

FINAL REPORT YANKEE MINE SITE UTAH COUNTY, UTAH **AUGUST 2002**

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1.0 INTRODUCTION

1.1 Objective of Study

The objective of this project was to conduct an extent of contamination investigation at the Yankee Mine, Globe Mine, Scotchman, and Plume L-94 mine tailings sites located in the Utah County, Utah. The investigation used field screening methodologies [X-Ray Fluorescence (XRF)] to assess metal contaminant levels (e.g., lead, zinc, and arsenic) in surface and subsurface soils on and surrounding the tailings pile associated with the sites' past mining operations. Surface water samples were collected from Mary Ellen Gulch and from observed groundwater discharges (e.g., seeps and springs) or mine discharges.

1.2 Site Description and Background

The Yankee Mine site is located in central Utah in the Uinta National Forest, approximately 25 miles east of Orem, Utah (Figure 1). The site is located on the southern side of the Wasatch Mountain Range between Twin Peaks and Mount Baldy at elevations ranging from 7,700 to 9,500 feet. The area is rural with very limited access by dirt roads or trails. The Yankee Mine, Globe Mine, and Plume L-94 are located near the headwaters of Mary Ellen Gulch. The Scotchman site is located further down the valley, adjacent to the American Fork Stream. Numerous mining operations utilized the area until the early to mid-1900s. Residual tailings piles from these mining operations are located throughout the American Fork Canyon watershed. Previous analytical results from the sampling of other tailing piles in the area have indicated the presence of heavy metals (including lead, arsenic, and zinc). In an effort to minimize the impact on public health and the environment posed by contaminants in the tailings, a Federal project is planned to select a nearby location to be used as a repository for the mine tailings from selected sites.

The United States Environmental Protection Agency/Environmental Response Team Center (U.S. EPA/ERTC) has been requested by the U.S. EPA Region VIII On Scene Coordinator (OSC) to perform a field investigation to meet the above objectives. Response Engineering and Analytical Contract (REAC) personnel conducted this investigation utilizing the technical approach described below.

2.0 METHODOLOGY

2.1 Surface Water Sampling

A total of fourteen surface water samples were collected from Mary Ellen Gulch, springs, seeps, and mine drainages observed on the site. When sufficient water was present, surface water samples were collected directly into appropriately labeled sample containers. Where low levels of water were present, surface water was transferred to a sample container using a dedicated plastic scoop. All sampling was performed per ERT/REAC Standard Operating Procedure (SOP) #2013, Surface Water Sampling and all site activities were documented in a site log book (Appendix A). Water samples were preserved by adding 40 percent nitric acid (trace metal grade) until the pH was below 2. Water samples for Target Analyte List (TAL) metals were maintained and shipped to the REAC Laboratories in Edison, NJ for TAL metal analysis on wet ice at approximately 4 degrees Celsius (°C)

At each sample location, water quality parameters (pH, conductivity, dissolved oxygen, and temperature) were collected using a Horiba U10 Water Quality Analyzer. Data were recorded in the site log book.

2.2 Soil Sampling

A total of ninety eight surface and seven subsurface soil samples were collected to evaluate the extent of contamination. Surface soil samples were collected from the upper 4-inches of soil using a dedicated stainless steel spoon or plastic scoop. Samples were placed in an 8-ounce glass jar or a self-sealing plastic bag, transported to the staging area, and thoroughly homogenized. After mixing, aliquots were taken for XRF analysis. For confirmation of the field screening analysis, 28 XRF cups (29 percent) were submitted to the REAC Laboratory for TAL metal analyses using graphite furnace atomic absorption (GFAA) and inductively-coupled plasma (ICP).

Subsurface samples were collected at three locations to evaluate the depth of the tailings and the extent of contamination. The intention was to collect samples at 1-foot intervals to a depth of 6-feet below ground surface (bgs), but refusal was encountered at much shallower depths (1 to 4 feet bgs) in all cases. Subsurface samples were collected using a bucket auger to dig to the specified depth, then a second decontaminated bucket auger was used to collect the sample. The sample material was transferred from the bucket auger to an 8-ounce glass jar and transported to the staging area for XRF analysis. Confirmation samples were shipped on wet ice at a temperature of approximately 4°C to the REAC Laboratory for TAL metal analysis.

2.3 X-Ray Fluorescence Analysis

All soil samples were screened in the field for zinc, arsenic and lead using a Spectrace 9000 field portable XRF analyzer (Appendix B). Samples were prepared by sieving homogenized soil through a 20-mesh sieve and transferring the sieved soil to XRF cups. XRF analysis was conducted in accordance with ERT/REAC SOP #1713, Spectrace 9000 Field Portable X-Ray Fluorescence Operating Procedure. Field screening data was provided to the U.S. EPA/ERTC Work Assignment Manager (WAM) daily, and this data was used to select additional sampling areas.

Regression analysis was used to compare XRF and laboratory confirmation data as specified in the U.S. EPA/ERTC Quality Assurance Technical Bulletin. If there is a significant relationship [coefficient of determination $(r^2) \ge 0.70$] between XRF and laboratory confirmation data, the XRF data will be considered Quality Assurance Level 2 (QA2) data. If a significant relationship does not exist $(r^2 < 0.70)$, the data will be considered Quality Assurance Level 1 (QA1) or screening data.

2.4 Site Maps

During this site investigation, all sample locations and site features (including Mary Ellen Gulch and its tributaries) were mapped using a Trimble PRO-XRS real-time differential Global Positioning System (GPS) equipment. The accuracy of the GPS varies depending on field conditions, but is typically accurate to within 1 to 5 meters. These data were used

to construct maps of site features and sample locations and were also overlaid on United States Geological Survey (USGS) 1-meter digital ortho-corrected aerial photographs.

2.5 Standard Operating Procedures

Documentation was conducted in accordance with the following standard operating procedures (SOPs):

REAC SOP #2002, Sample Documentation REAC SOP #4001, Logbook Documentation REAC SOP #4005, Chain of Custody Procedures

Sample packaging, packaging, shipment, storage, preservation, and handling were conducted in accordance with the following SOPs:

REAC SOP #2003, Sample Storage, Preservation, and Handling REAC SOP #2004, Sample Packaging and Shipment

Field sampling and on-site analytical techniques were conducted in accordance with the following SOPs:

ERT/REAC SOP #1713, Spectrace 9000 Field Portable X-Ray Fluorescence
Operating Procedure

ERT/REAC SOP #2001, General Field Sampling Guidelines
ERT/REAC SOP #2005, Quality Assurance/Quality Control Samples
ERT/REAC SOP #2006, Sampling Equipment Decontamination
ERT/REAC SOP #2012, Soil Sampling
ERT/REAC SOP #2013, Surface Water Sampling

3.0 RESULTS AND DISCUSSION

All soil samples collected were screened for lead (Pb), arsenic (As), and zinc (Zn) in the field by XRF (Table 1 for surface soil data, Table 2 for subsurface soil data and Appendix B for the XRF Final Report). Twenty-eight of these soil samples (25 locations and 3 duplicate samples) were submitted for TAL metals confirmation analysis (Table 3). Samples selected for confirmation analysis included a full range of contaminant concentrations. Fifteen surface water samples were collected (14 samples and 1 duplicate sample) and analyzed for TAL metals. The results in micrograms per liter (μ g/L) are summarized in Table 5.

For all inorganic laboratory data, metal concentrations below the method detection limit (MDL) were reported on the tables as not detected (ND). Appendices B (XRF Final Report) and C (Final Analytical Report), contain the supporting analytical reports, which provide a complete summary of all results, detection limits, and methods.

3.1 XRF Field Screening

As summarized in Table 4, XRF performed well as a screening tool for quantifying arsenic and zinc across the entire concentration gradient and lead at concentrations below 10,000

milligrams per kilogram (mg/kg). The coefficient of determination (r^2) between XRF and laboratory confirmation data exceeded 0.70 for both As ($r^2 = 0.73$) and Zn ($r^2 = 0.96$), indicating that the XRF data is considered QA-2 level data. Using all data points, there was a poor correlation for Pb between laboratory and XRF data ($r^2 = 0.45$). This appeared to be a result of low XRF response for concentrations exceeding 10,000 mg/kg. If all data points exceeding 10,000 mg/kg for Pb were excluded in the regression analysis, a higher coefficient of determination ($r^2 = 0.71$) was attained. Therefore, samples with XRF Pb concentrations below 10,000 mg/kg are considered to be QA2 level data, while samples with XRF Pb concentrations above 10,000 mg/kg are considered to be QA1 level data (screening data).

All samples with XRF Pb concentrations greater than 10,000 mg/kg are expected to have higher Pb concentrations than quantified by XRF. The x-coefficients in the regression analyses show that the XRF data underestimated metal concentrations compared to the laboratory analyses. X-coefficients were 1.2 for As, 2.0 for Pb and 2.1 for Zn. Multiplying the x-coefficient by the XRF concentration would provide a more comparable concentration to the fixed laboratory based data. The XRF MDLs for Zn, As, and Pb were 102 mg/kg, 72 mg/kg, and 41 mg/kg, respectively. Due to the Pb to As ratio, As concentrations were not quantified below one-tenth of the Pb concentration.

When evaluating this data, the inaccuracy of XRF lead data in samples at concentrations above 10,000 mg/kg must be emphasized. The highest Pb concentration quantified by XRF was 18,000 mg/kg, whereas this sample quantified by the laboratory analysis was 41,000 mg/kg. Ten of the 28 samples submitted for confirmation had Pb concentrations exceeding 18,000 mg/kg with the highest concentration being 95,000 mg/kg at location 61 (Table 4). Although the XRF provided accurate data for samples below 10,000 mg/kg of Pb, it was not able to accurately quantify Pb concentrations in samples exceeding 10,000 mg/kg.

It should also be noted that there is a moderate interference between As and Pb on the XRF. Arsenic will not be detected at concentrations below one-tenth of the Pb concentration in the sample. For example, if Pb was detected at 16,000 mg/kg in a sample, As would only be quantified if the concentration exceeded 1,600 mg/kg. This interference caused As to appear to be detected less frequently (i.e., more non-detects) then it may actually occur at the site. In the samples sent to the fixed laboratory for confirmation analysis, As was reported as not detected by XRF in 14 of the 28 samples. These same samples analyzed in the laboratory had As concentrations ranging from 120 mg/kg to 680 mg/kg with an average concentration of 384 mg/kg.

3.2 Surface Water

Samples SW-1, SW-7, SW-9, SW-10, and SW-12 were collected directly from Mary Ellen Gulch (Figure 2). Sample SW-11 was collected from an unnamed gulch that drains the northeastern portion of the valley and is not impacted by the Yankee Mine, Globe Mine and Plume L-94 tailings piles. This unnamed gulch joins Mary Ellen Gulch down gradient of the tailings pile and between locations SW-10 and SW-12. Samples SW-2, SW-3, SW-4, SW-5, and SW-6 were collected from or immediately downstream of spring or mine discharges along a hillside that was historically mined. SW-2 was collected from a spring discharge up gradient of the southern Yankee Mine tailings pile. The spring had a flow of

about 5 gallons per minute (gpm) and ran down the hillside for a short distance before flowing into the tailings. SW-3 was collected from a spring discharge with a flow of less then 0.5 gpm. The water discharged from the spring flowed across the ground and into a small bermed pond on top of the tailings pile. The actual sample was collected from a small pool of water between the spring and the pond. Sample SW-4 was collected from a small pool between the mine adit and the pond, formed by discharges from a mine adit with a flow rate of less then 0.5 gpm. The water traveled across the ground and into the same bermed pond where sample SW-3 was taken. The berm on the pond was breached on the south side and water from the pond flowed through the breach and onto the tailings pile. Samples SW-5 and SW-6 were both collected from spring discharges near the east edge of the Yankee Mine (north) tailings pile. The spring discharge where sample SW-5 was taken was greater then 5 gpm while the discharge where SW-6 was taken was less then 0.5 gpm. Water from both of these springs flowed across the tailings/access road and formed three separate drainage pathways before flowing over the edge of the slope and joining Mary Ellen Gulch. SW-8 was collected from one of these drainage pathways just before its confluence with Mary Ellen Gulch. Seep-1 and Seep-2 were collected from two seeps located between the southern extent of the Yankee Mine tailings pile and Mary Ellen Gulch. Both had flow rates below 1 gpm and water from the seeps flowed a short distance on the ground before discharging into Mary Ellen Gulch. Sample locations are indicated on Figure 2 and water quality data are summarized in Tables 5 (dissolved surface water metal concentrations) and 6 (other surface water quality parameters).

The upstream reference sample contained As, Pb, and mercury (Hg) at concentrations below the analytical MDLs and copper (Cu) and Zn at 44 micrograms per liter (μ g/L) each. Sample SW-11 contained As, Pb, Hg, Cu, and Zn at concentrations below the analytical MDLs.

With the exception of SW-5, all seeps and mine discharges (SW-2, SW-3, SW-4, SW-6) up gradient of the Yankee Mine tailings pile contained metal concentrations near or below concentrations found at the reference location (Figure 3). The sample collected at SW-5 contained elevated levels, compared to the reference location, of Zn (290 μ g/L) and As (83 μ g/L). Water from these discharges, with the exception of SW-5, did not seem to contribute a significant source of metal contamination to Mary Ellen Gulch.

The source of the water for SW-8 was the same as the source for SW-5 and SW-6, but SW-8 was collected approximately 150 feet further downstream, after the water flowed across the parking lot (tailings) and down most of the face of the tailings pile, but immediately before its confluence with Mary Ellen Gulch. SW-8 had much higher levels of As (920 μ g/L), Cu (110 μ g/L), Pb (230 μ g/L), and Zn (820 μ g/L) than either SW-5 or SW-6. This suggests that significant quantities of metals are dissolved in the surface water, after traveling over this portion of the tailings pile.

Samples collected from SEEP-1 and SEEP-2 contained greater than 10 times the levels of Zn (670 μ g/L and 770 μ g/L, respectively) compared to the reference sample (SW-1). Most likely these seeps travel (at least part of the way) through the tailings before discharging down gradient of the tailings pile.

Five surface water samples (SW-1, SW-7, SW-9, SW-10, and SW-12) were collected directly from Mary Ellen Gulch. The reference sample, SW-1, was collected upstream and up gradient of the Globe tailings pile. The next sample, SW-7 was collected downstream of the Globe Mine tailings, but upstream of the Yankee Mine tailings. Three more samples were collected from Mary Ellen Gulch: SW-12 was immediately up gradient of the confluence of Mary Ellen Gulch and an unnamed Gulch, SW 10 was collected down gradient of the confluence, and SW-9 was collected furthest downstream and immediately upstream of the split in the gulch. The samples collected further downstream were impacted by an increased number of surface and subsurface drainage flows that came in contact with the tailings. As shown in Figure 4, there was an increasing trend in metal concentrations in Mary Ellen Gulch.

3.3 Soil Samples

Elevated concentrations of Zn and Pb were associated with most samples collected on and adjacent to the Yankee Mine, Globe Mine, Plume L-94, and Scotchman tailings piles (Table 1, 2, and 3 and Figure 5). The discussion below compares XRF data between locations (only XRF data was collected at all sample locations).

Overall there were no patterns or trends in metal concentrations within each tailings pile (Figure 6, 7, and 8). Metal concentrations in each tailings pile were highly variable. Table 7 contains a summary of mean (x) concentrations for Zn, As, and Pb in each of the tailings piles. The highest average concentration of Zn ($\bar{x} = 4,328 \text{ mg/kg}$) were taken in the L-94 tailings pile. As $(\bar{x} = 194 \text{ mg/kg})$ in the Yankee Mine (north) tailings pile, and Pb $(\bar{x} = 3.543)$ mg/kg) in the Globe tailings pile. The standard errors indicate the high degree of variability within each pile. Several additional soil samples were taken from the tailings pile. Four samples (87, 88, 89, 90) were collected below a bermed pond down gradient of the Yankee (south) tailings pile. The berm appeared to be constructed from tailings and there was a breach in the center of it, allowing water to overflow at high water levels. These samples contained levels of Zn ranging from 1,900 mg/kg to 21,000 mg/kg and Pb ranging from 3,600 mg/kg to 4,800 mg/kg. Arsenic was not detected in any of these samples. Nine soil samples (31, 32, 51 through 56, and 94) were collected off the tailings pile adjacent to Mary Ellen Gulch. These samples contained Zn levels ranging from 420 mg/kg to 3,200 mg/kg. As concentrations ranging from not detected (ND) to 950 mg/kg (second highest As concentration of all samples collected on the site), and Pb concentrations ranging from 230 mg/kg to 9,100 mg/kg. Location 52 (Zn = 2,200 mg/kg, Pb = 2,500 mg/kg) appeared to have been used as a campsite.

Subsurface samples were collected at three locations, but depths of sample collection was limited due to refusal (Table 2). Between the three locations, there were no overall patterns of contamination relative to depth. At location 9, the highest Zn and Pb concentrations were detected at a depth of 2 feet bgs and the highest As concentration was detected at a depth of 1 foot bgs. At locations 11 and 96, the highest Zn and Pb concentrations were detected at the surface and the highest As concentration was detected at a depth of 1-foot.

4.0 CONCLUSIONS

Lead, arsenic, and zinc were detected at concentrations up to 95,000 mg/kg (based on laboratory confirmation data), 16,000 mg/kg (based on XRF data), and 59,000 mg/kg (based on laboratory confirmation data), respectively, in the tailings and soil samples collected at the site. The contamination wasn't limited to specific portions of the site but was widespread both on and off the tailings piles. Several samples collected off of the tailings piles adjacent to Mary Ellen Gulch (including one location which appeared to be used as a camp site) contained elevated Pb concentrations (up to 21,000 mg/kg based on confirmation data), Zn (up to 6,700 mg/kg based on confirmation data), and As (up to 950 mg/kg based on XRF data). When comparing tailings piles, plume L-94 had the highest average Zn concentration ($\bar{x} = 4,328$ mg/kg), Yankee Mine (north) had the highest average As concentration ($\bar{x} = 194$ mg/kg), and Globe Mine had the highest average Pb concentration ($\bar{x} = 3,543$ mg/kg).

Five of the six springs/mine discharges sampled up gradient of the Yankee Mine tailing piles had comparable contaminant levels to the reference sample, while SW-05 contained elevated levels of Zn and As. The water source for samples SW-8 (discharge 5 and 6) was the same as the sources for SW-5 and SW-6, but SW-8 was collected approximately 150 feet downstream after flowing over tailings. SW-8 contained much higher As, Cu, Pb, and Zn concentrations compared to SW-5 and SW-6. This provides evidence that the tailings are a source of metal contamination to the Mary Ellen Gulch. In addition, two seep samples (SEEP-1 and SEEP-2) collected down gradient of the Yankee Mine tailings pile also contained elevated levels of Zn and Cu compared to the spring samples (SW-2, SW-3, SW-4, and SW-6) collected up gradient of the tailings pile. Several surface water samples collected from Mary Ellen Gulch, starting up gradient of the Globe Mine tailings pile, indicated increasing metals concentrations after flowing down gradient across the Globe Mine tailings and again after flowing down gradient of the Yankee Mine tailings piles.

Tables

Table 1 XRF Metals in Surface Soils Yankee Mine Site Utah County, Utah August 2002

All Concentrations are mg/kg (ppm)

Sample	Date	Zinc	Arsenic	Lead
1	2 October 2001	630	150 J	800
1 (DUP)	2 October 2001	780	150 J	610
2	2 October 2001	5,400	240 J	4,300
2 (REP)	2 October 2001	1,900	ND	4,500
2 (DUP)	2 October 2001	1,900	ND ND	5,000
3	2 October 2001	530	ND ND	2,100
3 (DUP1)	2 October 2001	830	ND	2,200
3 (DUP2)	2 October 2001	520	ND ND	2,000
3 (DOFZ)	2 October 2001	4,700	ND ND	8,800
4 (DUP)	2 October 2001	5,400	ND ND	8,300
	2 October 2001		ND ND	4,200
5		1,800		
5 (DUP)	2 October 2001	1,400	ND	3,600
6	2 October 2001	890	ND	1,700
6 (DUP)	2 October 2001	1,000	ND	1,800
<u>/</u>	2 October 2001	2,600	ND	7,200
7 (DUP)	2 October 2001	3,000	ND	7,200
8	2 October 2001	200 J	ND	88 J
8 (DUP)	2 October 2001	200 J	ND	98 J
9	2 October 2001	420	150 J	480
9 (DUP)	2 October 2001	240 J	140 J	470
10	2 October 2001	2,800	ND	1,600
10 (DUP)	2 October 2001	2,600	180 J	1,400
11	2 October 2001	ND	ND -	1,600
11 (DUP)	2 October 2001	110 J	ND	1,900
12	2 October 2001	1,800	540	520
13	2 October 2001	290 J	ND	1,100
14	2 October 2001	990	220 J	870
15	2 October 2001	600	100 J	180
16	2 October 2001	680	150 J	320
17	2 October 2001	560	ND	2,000
18	2 October 2001	7,500	ND	6,100
18 (DUP)	2 October 2001	6,800	ND	5,400
19	2 October 2001	160 J	ND	1,100
20	2 October 2001	2,100	ND	4,600
21	3 October 2001	ND	ND	260
22	3 October 2001	330 J	130 J	1,200
22 23	3 October 2001	1,000	350	1,700
23 (DUP)	3 October 2001	620	190 J	1,700
24	3 October 2001	510	ND	2,100
24 25 26 27	3 October 2001	1,100	ND	4,500
26	3 October 2001	730	ND	2,600
27	3 October 2001	1,200	ND	920
28	3 October 2001	8,800	ND	13,000
28 (DUP)	3 October 2001	11,000	ND	15,000
29	3 October 2001	28,000	ND ND	16,000
30	3 October 2001	1,300	370	1,600
		ins above the deta		1,000

ND = Not Detected at concentrations above the detection limit.

J = Estimated Value mg/kg = milligrams/kilogram ppm = parts per million

Table 1 (Cont'd) XRF Metals in Surface Soils Yankee Mine Site Utah County, Utah August 2002

All Concentrations are mg/kg (ppm)

Sample	Date	Zinc	Arsenic	Lead
31	3 October 2001	790	340	2,700
32	3 October 2001	11,000	ND	14,000
33	3 October 2001	200 J	650	1,200
34	3 October 2001	680	410	2,200
35	3 October 2001	2,800	600	4,000
36	3 October 2001	3,600	ND	3,700
37	3 October 2001	1,700	350	2,200
38	3 October 2001	250 J	ND	1,400
39 (DUP)	3 October 2001	650	ND	3,800
40	3 October 2001	2,200	ND	2,600
41 (DUP)	3 October 2001	160 J	130 J	670
42	3 October 2001	120 J	ND	2,400
43	3 October 2001	130 J	170 J	690
43 (DUP)	3 October 2001	310 J	130 J	870
44	3 October 2001	310 J	1,600	410
45	3 October 2001	760	240 J	920
46	3 October 2001	170 J	ND	2,400
47	3 October 2001	130 J	ND	2,000
48	3 October 2001	600	ND	4,800
49	3 October 2001	1,600	ND	2,900
50	3 October 2001	340	87 J	80 J
51	3 October 2001	1,100	ND	1,100
51 (DUP)	3 October 2001	1,100	ND	1,100
52	3 October 2001	2,100	ND	2,600
52 (DUP)	3 October 2001	2,300	ND	2,400
53	3 October 2001	420	ND	230
53 (DUP)	3 October 2001	420	ND	280
54	3 October 2001	1,500	950	6,800
54 (DUP)	3 October 2001	2,700	810	5,500
55	3 October 2001	3,200	500	2,200
56	3 October 2001	1,700	ND	9,100
56 (DUP)	3 October 2001	1,200	ND	7,800
56 (DUP/REP)	3 October 2001	1,400	ND	7,900
57	3 October 2001	1,600	270	830
58	3 October 2001	960	100 J	480
59	3 October 2001	280 J	300	85 J
60	3 October 2001	3,800	ND	1,700
61	3 October 2001	14,000	ND	12,000
61 (DUP)	3 October 2001	15,000	ND ND	13,000
62	3 October 2001	820	320	120 J
63	3 October 2001	610	ND ND	4,700
64	3 October 2001	13,000	ND /	16,000
64 65	3 October 2001	300 J	ND ND	830
66 66	3 October 2001	320 J	710	530
67	3 October 2001	300 J	650	290
68	3 October 2001	1,200	ND ND	4,400
	ted at concentration		<u> </u>	4,400

ND = Not Detected at concentrations above the detection limit.

J = Estimated Value mg/kg = milligrams/k:logram ppm = parts per million

Table 1 (Cont'd) XRF Metals in Surface Soils Yankee Mine Site Utah County, Utah August 2002

All Concentrations are mg/kg (ppm)

Sample	Date	Zinc	Arsenic	Lead
69	3 October 2001	870	470	160
70	3 October 2001	150 J	150 J	660
71	3 October 2001	820	ND	2,200
72	3 October 2001	660	420	1,200
73	3 October 2001	1,500	ND	630
74	3 October 2001	1,200	ND	3,300
75	3 October 2001	2,300	ND	9,800
75 (DUP)	3 October 2001	3,000	ND	11,000
76	3 October 2001	1,700	ND	4,800
77	3 October 2001	450	ND	4,900
78	3 October 2001	870	ND	4,600
79	3 October 2001	1,100	ND	4,900
80	3 October 2001	630	ND	5,600
81	3 October 2001	360	ND	3,900
82	3 October 2001	670	ND	4,000
83	3 October 2001	350	ND	1,600
84	3 October 2001	190 J	ND	2,700
85	3 October 2001	ND	ND	3,000
86	3 October 2001	140 J	120 J	900
86 (DUP)	3 October 2001	ND	ND	900
87	4 October 2001	1,100	ND	2,100
88	4 October 2001	1,500	380	430
89	4 October 2001	220 J	ND	ND
90	4 October 2001	370	87 J	210
91	4 October 2001	1,900	ND	4,500
91 (DUP)	4 October 2001	2,200	ND	4,700
92	4 October 2001	1,900	ND	3,600
93	4 October 2001	3,400	ND	4,200
93 (DUP)	4 October 2001	2,600	ND	4,000
94	4 October 2001	21,000	ND	4,800
95	4 October 2001	140 J	ND	45 J
96	4 October 2001	650	ND	1,900
97	4 October 2001	2,400	ND	8,700
98	4 October 2001	560	ND	330

ND = Not Detected at concentrations above the detection limit.

J = Estimated Value mg/kg = milligrams/kilogram ppm = parts per million

Table 2 XRF Metals in Subsurface Soils Yankee Mine Site Utah County, Utah August 2002

All Concentrations are mg/kg (ppm)

Sample	Date	Zinc	Arsenic	Lead
9	2 October 2001	420	150 J	480
9 (DUP)	2 October 2001	240 J	140 J	470
9-1'	4 October 2001	470	_ 810	3,000
9-2'	4 October 2001	6,100	ND	18,000
9-3' 9-4'	4 October 2001	5,900	ND	14,000
	4 October 2001	430	740	360
11	2 October 2001	ND	ND	1,600
11 (DUP)	2 October 2001	110 J	ND	1,900
11-1'	4 October 2001	ND	_ 110 J	800
11-2'	4 October 2001	ND	ND	1,000
96	4 October 2001	650	ND	1,900
96-1'	4 October 2001	210 J	89 J	110 J

ND = Not Detected at concentrations above the detection limit.

J = Estimated Value mg/kg = milligrams/kilogram ppm = parts per million

Table 3
TAL Metals in Soils
Yankee Mine Site
Utah County, Utah
August 2002

All Concentrations are mg/kg (ppm)

Location	Arsenic	Copper	Lead	Mercury	Zinc
2	650	500	11,000	14	3,400
4	430	79	16,000	47	14,000
7	490	280	20,000	4.4	5,600
10	120	340	2,600	3.7	4,700
10 (DUP)	170	220	2,300	3.2	4,700
12	360	280	560	0.64	1,500
14	290	1,100	880	0.52	1,600
18	680	470	20,000	39	11,000
20	140	15,000	4,800	2.1	1,500
23	300	700	1,700	1.7	1,300
23 (DUP)	300	680	1,800	1.1	920
25	400	2,300	3,800	1.7	1,100
28	370	330	32,000	51	23,000
29	280	340	21,000	67	59,000
32	270	310	17,000	35	21,000
35	1,200	890	5,200	4.2	4,400
44	1,500	67	340	0.26	160
50	43	50	86	0.34	240
54 (DUP)	820	760	21,000	4.8	6,700
54	820	1,500	25,000	5.4	5,100
59	230	36	95	0.15	190
61	560	3,000	95,000	3.0	29,000
64	520	3,600	23,000	32	20,000
75	160	1,600	30,000	2.1	5,700
86	81	37	930	0.3	46
97	300	550	11,000	4.7	2,600
9-2'	660	990	41,000	72	13,000
9-4'	560	170	310	0.42	360

mg/kg = milligrams/kilogram ppm = parts per million

Table 4 Comparison of XRF and TAL Metal Concentrations Yankee Mine Site Utah County, Utah August 2002

All Concentrations are mg/kg (ppm)

	ARSI	ENIC	LE	AD	ZINC		
Location	TAL	XRF	TAL	XRF	TAL	XRF	
2*	650	160 J	11,000	4,400	3,400	3,700	
4	430	ND	16,000	8,800	14,000_	4,700	
7	490	ND	20,000	7,200	5,600	2,600	
10	120	ND_	2,600	1,600	4,700	2,800	
10 (DUP)	170	180 J	2,300	1,400	4,700	2,600	
12	360	540	560	520	1,500	1,800	
14	290	220 J	880	870	1,600	990	
18	680	ND	20,000	6,100	11,000	7,500	
20	140	ND	4,800	4,600	1,500	2,100	
23	300	350	1,700	1,700	1,300	1,000	
23 (DUP)	300	190 J	1,800	1,700	920	620	
25	400	ND	3,800	4,500	1,100	1,100	
28	370	ND	32,000	13,000	23,000	8,800	
29 32	280	ND	21,000	16,000	59,000	28,000	
32	270	ND	17,000	14,000	21,000	11,000	
35	1,200	600	5,200	4,000	4,400	2,800	
44	1,500	1,600	340	410	160	310 J	
50	43	87 J	86	80 J	240	340	
54	820	950	21,000	6,800	6,700	1,500	
54 (DUP)	820	810_	25,000	5,500	5,100	2,700	
59	230	300	95	85 J	190	280 J	
61	560	ND	95,000	12,000	29,000	14,000	
64	520	ND	23,000	16,000	20,000	13,000	
75	160	ND	30,000	9,800	5,700	2,300	
86	81	120 J	930	900	46	140 J	
97	300	ND	11,000	8,700	2,600	2,400	
9-2'	660	ND	41,000	18,000	13,000	6,100	
9-4'	560	740	310	360	360	430	
coefficient of correlation	0.		0.45* 0.71**		0.96		
x-coefficient	1.	2	2	.0	2	1	

ND = Not Detected at concentrations above the detection limit.

J = Estimated Value mg/kg = milligrams/kilogram ppm = parts per million

* = All Data points

^{** =} data points < 10,000 mg/kg

Table 5 Dissolved TAL Metals in Surface Waters Yankee Mine Site Utah County, Utah August 2002

All Concentrations are µg/L (ppb)

SW-01	SW-01 DUP	SW-02	SW-03	SW-04	SW-05	SW-06	SW-07	SW-08	SW-09	\$W-10	ŚW-11	SW-12	SEEP 1	SEEP 2
480	480	ND	340	260	100	ND	140	3,300	1,600	560	ND	520	ND	100
ND	ND	ND	ND	ND	2.9	ND	4.6	46	13	6.5	ND	13	2.8	NĎ
ND	ND	OND	2.8	4.0	83	ND	ND	920	98	49	ND	63	8.4	ND:
23	23	61	79	24	15	78	26	67	68	36	39	32	150	66
ND	ND	ND	ND	ND	ND	ND	ND	2.3	ND	ND	ND	ND	ND	ND
ND	ND	ND	ND	ND	ND	ND	ND	ND	7.4	ND	ND	ND	ND	5.0
20,000	20,000	37,000	48,000	64,000	23,000	44,000	24,000	25,000	30,000	29,000	33,000	27,000	47,000	44,000
ND_	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
ND	ND	NĎ	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
44	45	ND	ND	ND	ND	NO	12	110	140	37	ND	47	ND	63
37	39	ND	400	_590	8,200	77	250	120,000	14,000	5,700	26	7,300	8,500	99
ND	ND	ND	11	12	2.9	5.2	6.0	230	180	57	ND	130	NĎ	11
7,700	7,800	19,000	25,000	33,000	8,800	15,000	9,600	9,700	12,000	11,000	11,000	11,000	23,000	22,000
16	17	ND	25	29	140	5.2	19	200	110	79	ND	100	380	85
ND	ND	ND	ND	ND	NO NO	ND	ND	ND	0.28	ND	ND	ND	ND	ND
ND	ND	ND	ND	ND	12	ND	ND	16	12	ND	ND	ND	ND	ND
ND	ND	ND	ND	ND	NO	NĐ	ND	ND	2,500	ND	ND	ND	ND	ND
ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
ND	ND	ND	ND	ND	ND	NO	ND	ND	ND	ND	ND	ND	ND	ND
770	750	750	820	900	880	1,400	770	900	990	870	900	850	1,100	840
ND	ND	ND	ND	ND	, ND	ND	ND	86	ND	ND	ND	ND	ND	ND
ND	ND	ND	ND	ND	ND	ND	ND	ND	. NO	NĐ	ND	ND	ND	ND
44	44	26	21	34	290	58	160	820	1,000	320	ND	420	670	770
	480 ND ND ND 23 ND ND 20,000 ND ND 44 37 ND 7,700 16 ND	480 480 ND ND ND ND ND ND 23 23 23 ND ND ND ND ND ND ND ND ND ND ND ND ND 44 45 37 39 ND ND 7,700 7,800 16 17 ND ND ND ND	480 480 ND N	480	A80	A80	A80	A80	A80	A80	A80	A80	480	A80

ND = Not Detected at concentrations above the detection limit.

ug/L = micrograms/Liter

ppb = parts per billion

Table 6 Surface Water Qualitity Paramters Yankee Mine Site Utah County, Utah August 2002

		Dissolved			Estimated
	pН	Oxygen	Temperature	Conductivity	Flow Rate
Location	(units)	(mg/L)	(C)	(mS/cm)	(gallons/minute)
SW-1	NA	NA NA	NA	NA	NA NA
SW-2	7.83	11.30	7.9	0.600	> 5
\$W-3	8.12	7.89	13.9	0.509	< 0.5
SW-4	8.20	8.50	12.8	0.596	< 0.5
SW-5	6.72	7.85	7.2	0.228	> 5
SW-6	7.10	5.49	13.2	0.324	< 0.5
SW-7	7.65	9.29	7.1	0.203	10 - 15
SW-8	7.49	9.20	7.7	0.013	NÄ
SW-9	7.97	8.74	9.3	0.244	> 25
SW-10	7.95	8.86	8.6	0.242	50 - 75
SW-11	8.11	8.23	10.9	0.246	20 - 30
SW-12	8.00	8.99	7.0	0.239	35 - 40
SEEP-1	7.29	4.20	10.4	0.414	< 1
SEEP-2	8.03	9.72	7.4	0.384	< 0.5

mg/L = milligrams/Liter C - degrees Celsius

mS/cm - millisiemens per centimeter

NA - Not Available

Table 7 Summary of XRF Metal Concentrations by Area Yankee Mine Site Utah County, Utah August 2002

All Concentrations are mg/kg (ppm)

		nc				
Location	Mean	Standard Deviation	Standard Error	Minimum	Maximum	Total Number of Samples
Global	1,284	2,564	200%	ND	13,000	24
L-94	4,328	6,110	141%	280	15,000	5
Scotchman	938	1,000	107%	140	2,400	4
Yankee North	1,984	4,526	228%	ND	28,000	41
Yankee South	1,348	2,029	151%	160	7,200	11

	Arsenic					
Location	Mean	Standard Deviation	Standard Error	Minimum	Maximum	Total Number of Samples
Global	153	204	134%	ND	710	24
L-94	150	126	84%	ND	300	5
Scotchman	ND	NA	NA	ND	ND	4
Yankee North	194	277	143%	ND	1,600	41
Yankee South	102	99	97%	ND	380	11

ř – – E	Lead					
Location	Mean	Standard Deviation	Standard Error	Minimum	Maximum	Total Number of Samples
Global	3,543	3,545	100%	120	16,000	24
L-94	3,219	5,500	171%	85	13,000	5
Scotchman	2,744	4,054	148%	45	8,700	4
Yankee North	2,891	3,316	115%	93	16,000	41
Yankee South	1,531	1,975	129%	ND	5,800	11

ND = Not Detected at concentrations above the detection limit.

mg/kg = milligrams/kilogram

ppm = parts per million

Figures

EPA REGION VIII SUPERFUND DOCUMENT MANAGEMENT SYSTEM

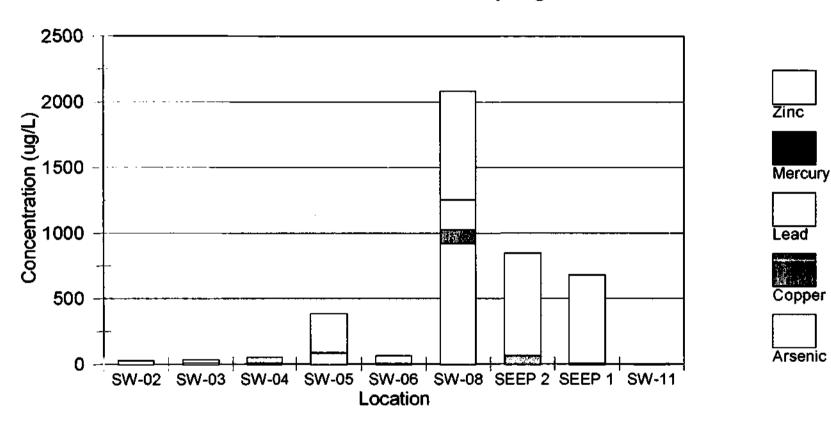
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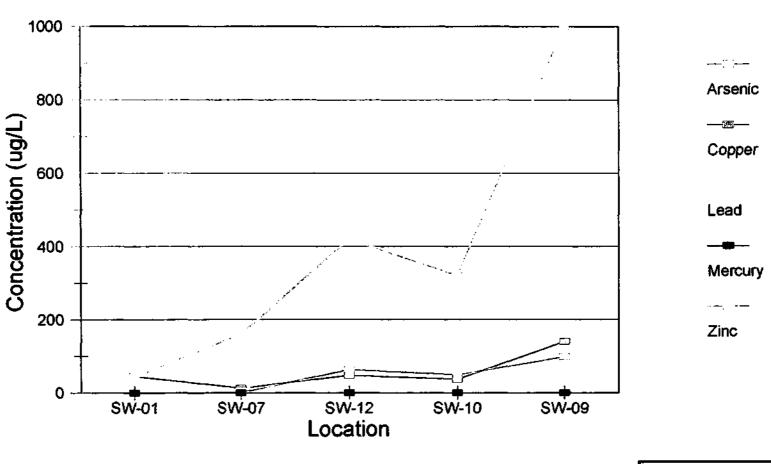
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	FIGURE 2 WATER SAMPLE LOACTIONS YANKEE MINE SITE UTAH CO., UTAH

Comparison of Metal Concentrations in Mine Adits and Springs



U.S. EPA Environmetal Response Team Center Response Engineering and Analytical Contract 68-C99-223 W.A. # R1A00232 Figure 3 Metals in Mine Discharges and Springs Yankee Mine Site Utah Co., Utah July 2002

Surface Water Metal Concentrations in Mary Ellen Gulch



U.S. EPA Environmetal Response Team Center Response Engineering and Analytical Contract 68-C99-223 W.A. # R1A00232 Figure 4
Comparison of Metal
Concentrations in Mine Discharges,
Seeps, and Springs
Yankee Mine Site
Utah Co., Utah
August 2002

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FIGURE 5 SOI	L SAMPLE LOCATION MAP
	

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	FIGURE 6 XRF LEAD CONCENTRATION IN SOILS

EPA REGION VIII SUPERFUND DOCUMENT MANAGEMENT SYSTEM

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FIGURE 7 XRF	ZINC CONCENTRATION IN SOILS

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FIGURE 8 ZRI	ARSENIC CONCENTRATIONS IN SOILS

Appendix A

Appendix A
Field Documentation
Yankee Mine Site
Unita National Forest, Utah
August 2002

M Cont case RVA00 232 Vantee Mine Site 10/1/01 Such Enoseman REAC
Larry Kapelin REAC
Royanna Winsman REAC
Alan Homphries ERT
Rude Shephenson Region 8 CSC
(SIV) Larpus 5 AR Surface Sarphantop Bindes
Lock - CRS Noor Sham
Cross rock 5:00 pax of
good 101/ings
Loc 2 5:04 PM M 5:04 5:04 5:04 5:04 5:04 5:04 5:04 5:04 5:04 5:04 5:10 5:10 5:10 5:10 5:10 5:10 5:10

10/2/0)
Little hadel 7:00AM
Amie Steel 7:45
Direct 1000 an Topologic
Transcet #1 W-E Topologic
Fine grain bet light gray North
to coarse gravel
North side Isste
TO 40.50 Ft W-E
P. le 20-30 Ft N-5 # 22 red/gray 1000
fine 51/t to correspond
7/00 ft down from top of pile # 23 1015 Large rock outcrop The grain sitt/course granul # 24 Med Gray 1020 Mootly Fine & Soft gravel # 25 10 Ft from Greek 1025 Lt brown fine grain organia / topsoi)

30 Ft La creek
50 Ft durstream of Confluence
steep gulley to creek ride buck up to staging area 1205 1207 # 33 Fine med brown Lane gravel area
erosion zone
75 how week new toe of large pile

Area B

Hard surface The med brown /gravel 1215 35 The med brown Crosion church

#42 1405 GPS 6' E of

Stk gray actual

Mechin size

Pile B - Main Pile

on face 50' from top Confacted, had surface fine suffered gravel fine of Pole B #44 Redush / Lt gray
Wiside of Pile B
50' h typ

45 th Gray
orange speeks
Wedge of Pile B
100' A top

1500 #46 Med gray
contact of h
have swfan

1645 S3 Form mill sity

1645 BV. dry silty topsol

1445 53 Form millsof Bern town willing 1650 34 10 x10 F7 pile new creek white piecep tailings ? 1715 #55 Red + gray | fine sitt to course next to crock 1720 It 56 Redish, coarse

Pile new to crack

Tailings was head de From mine? 1740 returned to thruck

of Sure John Sun my paper String Scotter Sun of 1972 String Scotter Scotter Sun of 1972 String Scotter brestion-techniques Soil

side of brank. Collocated

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ME C85 (USFS) Son draway creek -collocated.

Horiba 750 collocated an additional

with a son collocated an additional Wood pile-mid way + white gay

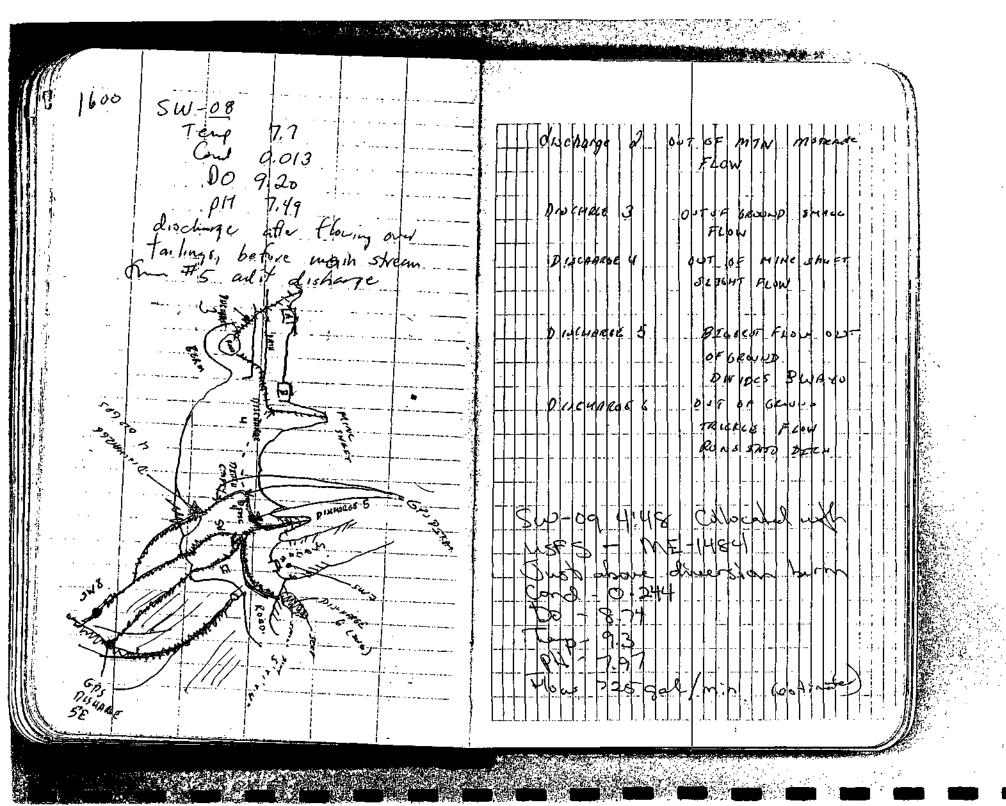
Fight Scot - Nand combod pret apa - En drain contect

5 - 11:55 Boyou bije. And From pile Just Sulow Land Sulf - on south Slope 20 - Same slope Surker downstream - brown - fine 25 Est 11:34 South By Sund Sind to Eq. Jarge saun

For Fine gours compart
bage 19119 win \$ 83 12:33 - Stay whitish Sursay Ind Stained Taling Down shope below word Sed 1200) Hot be Readings 1200) 1300 1200 500 13.00 13.00 13.00 54 Salpes Aladad in 1/2 Policy

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2 South State Sand State Sand State State State State State State Sand State Sand State Sand State Sand State Stat 06



10/3/01 4.20 104 Flow 1705 <u>6.242</u>

Deep coming indo latt 30 St down Sham Grond SW 12 and about \$30 SST upsham from the continued No real flow throbbe to called a sample Seep \$ 2 17:26 Comine of P. 4 B (Marrile) Draine She sis grasser grea ab Cost of give Flows 4/1/min Cond - 6.384 DD - 9.72 Texp- 7.4 PH- 8.03 15 ft downstreen Iron

Satchman Sudden Bond 18:50 South Brew go degrand Bond Brown of Browned 10/4/01 293 1810

B [~] A

Appendix B

Appendix B
XRF Final Trip Report Yankee Mine Site
Unita National Forest, Utah
August 2002

DATE:

16 August, 2002

TO:

Scott Grossman, REAC Task Leader

FROM:

Lawrence Kaelin, REAC Chemist/Acting Geology Group Leader

SUBJECT:

FPXRF ANALYSES, YANKEE MINE SITE, UNITA NATIONAL FOREST, UTAH

WORK ASSIGNMENT 0-232 - FINAL TRIP REPORT

BACKGROUND

A Spectrace 9000 Field-Portable X-ray Fluorescence (FPXRF) analyzer, maintained and operated by Response Engineering and Analytical Contract (REAC) personnel, was used to support United States Environmental Protection Agency/Environmental Response Team Center (U.S. EPA/ERTC) activities at the Yankee Mine site. REAC personnel analyzed site surface and subsurface soil samples for the primary target elements: zinc (Zn), arsenic (As), and lead (Pb).

REAC personnel were at the site from 02 to 04 October 2001, to determine the extent of target element contamination in site soil (mine tailings) samples utilizing a Spectrace 9000 FPXRF analyzer (S/N Q-011). Over 100 soil samples and duplicate samples were analyzed via FPXRF while on site. Twenty eight duplicate samples, including three (3) replicate samples, were sent to REAC, Edison, NJ, for confirmatory laboratory analyses.

OBSERVATIONS AND ACTIVITIES

Spectrace 9000 FPXRF Analyses

The Spectrace 9000 FPXRF measurement times (instrument live-time) were 60 seconds for the cadmium-109 (Cd-109) source, 60 seconds for the iron-55 (Fe-55) source, and 60 seconds for the americium-241 (Am-241) source.

Sample preparation, analysis, and quality assurance/quality control (QA/QC) procedures used in this study conform to those described in the U.S. EPA/ERTC REAC Standard Operating Procedure (SOP) #1713, Spectrace 9000 Field Portable X-ray Fluorescence Operating Procedure.

Preliminary results for target elements were reported on a daily basis during the site visit. All preliminary FPXRF on-site field results were QA-1 (screening) level data only. Confirmatory laboratory analyses were later determined on a subset of duplicate samples for the Target Analyte List (TAL) metals to raise the FPXRF results to QA-2 (definitive) level data.

Soil samples were received on site in labeled plastic bags or glass jars. Most samples were dry as received. Each sample was mixed with a spoon. Wet samples were prepared by placing 10-20 grams of the sample in a labeled aluminum weighing dish and oven drying for 1-2 hours as necessary. The dry sample was passed through a 10-mesh stainless steel sieve to remove rocks and large organic matter. The sample was then placed in a labeled 31-millimeter (mm) polyethylene X-ray sample cup and sealed with 0.2-mil (5 micrometer) thick polypropylene X-ray window film. Duplicates were prepared for approximately every 10th sample and the suffix "DUP or DU" was added to the sample ID for the duplicate sample. Prior to XRF analysis, each sample cup was tapped against the tabletop to pack the sample evenly against the film window. The sample cup was placed directly on the probe aperture window of the Spectrace 9000 FPXRF analyzer, the safety shield was closed, and analysis was initiated with the measurement times previously noted.

XRF analysis results for each sample were saved in the Spectrace 9000 internal data logger memory. The data were

downloaded and archived on computer disks on a daily basis. Target element results for each analyzed sample and standard were logged into the Spectrace 9000 field logbook. Target element results were qualified using the method detection and quantitation limits discussed below in this report. Table 1 summarizes the qualified FPXRF results.

OA/OC Procedures

The reliability of the Spectrace 9000 FPXRF unit and application model was evaluated daily during the site visit. The energy calibration check and detector resolution check were performed at the beginning of each day to ensure that proper instrument calibration was maintained and that the detector resolution was adequate for producing reliable X-ray intensity measurements. The Spectrace 9000 soil application model was verified at the beginning of each day for the target elements. This was accomplished by analyzing a blank sand sample (Mallinkrodt, Lot # 7062KJHP) and National Institute of Standards and Technology (NIST) Standard Reference Materials (SRMs) #2709, #2710, and #2711. Energy calibration checks, detector resolution checks, and application verification results were recorded in the Spectrace 9000 field logbook.

Method Detection and Quantitation Limits

A certified standard, NIST SRM #2709, was analyzed at the beginning of each day and periodically during sample analysis to establish statistically derived method detection and quantitation limits for the target elements. The standard deviation [STD (n-1)] for these analyses was used to calculate the Spectrace 9000 method detection limit (MDL) and method quantitation limit (MQL) for each target element. The MDL was calculated for each element as three times the standard deviation (MQL = 10 x STD) and the MQL was defined as ten times the standard deviation (MQL = 10 x STD) for repeat measurements of SRM #2709. The reported MDL was based on this calculation for both Pb and Zn. Lead interferes with the As analysis at Pb:As ratios of 5:1 or greater. Therefore, the reported As MDL was the statistical value or 1/10 the Pb concentration, whichever was greater. The MDLs and MQLs were calculated daily for the reporting of field data. An average MDL and MQL for the entire period from 02-04 October, 2001, was used to qualify the compiled field results in Table 1 and appears at the end of the table. Appendix A has the raw data used to calculate the daily and average MDLs and MQLs.

Spectrace 9000 results were qualified by a "U" for analyses with results less than the MDL (not detected). All preliminary FPXRF results determined in the field were QA-1 (screening) level data only. FPXRF results above the MDL but below the MQL were qualified by a "J" and represents results in a region of some statistical uncertainty. The daily field screening FPXRF results are in Appendix B.

FPXRF Confirmation Samples

In order to obtain a "definitive" QA-2 level for the FPXRF data set, a minimum of 10 percent of the field samples must be confirmed by a laboratory method such as Inductively-Coupled Plasma (ICP) emission spectroscopy or Atomic Absorption (AA) analysis. A regression analysis between the Spectrace 9000 data (independent) and the confirmatory data (dependent) must yield a coefficient of determination (r²) greater than 0.7 (U.S. EPA/ERT 1991). The model obtained by the regression may be used to validate or adjust the Spectrace 9000 data.

Twenty-eight (28) of the soil/sediment samples, including three (3) replicates, analyzed by FPXRF methods were selected by the Task Leader and submitted for confirmatory (TAL metals) laboratory analysis. To minimize potential sample homogeneity problems, the same XRF sample cups were submitted for confirmatory analysis. Confirmatory laboratory results for TAL metals are presented in Table 2.

Results

Table 1 contains qualified FPXRF results for target elements. Appendix A contains the MDL, MQL, linear regression data, and QA/QC data. Preliminary FPXRF field results for all samples are in Appendix B. Copies of field logbook entry pages are in Appendix C. Table 2 has the TAL metal confirmatory laboratory results. Table 3 compares the FPXRF and TAL confirmatory results for samples above the detection limits, along with a determination of their relative percent differences (RPD).

Lockheed Martin Technology Services
Environmental Services REAC
2890 Woodbridge Avenue Building 209 Annex
Edison, NJ 08837-3679
Telephone 732-321-4200 Facsimile 732-494-4021



DATE:

August 30, 2002

TO:

Alan Humphrey, U.S. EPA/ERTC Work Assignment Manager

THROUGH:

Gary Newhart, REAC Operations Section Leader

FROM:

Scott Grossman, REAC Task Leader

SUBJECT:

DOCUMENT TRANSMITTAL UNDER WORK ASSIGNMENT 0-232

Attached please find the following document prepared under this work assignment:

FINAL REPORT YANKEE MINE SITE UTAH COUNTY, UTAH

cc:

Central Files WA #0-232 (w/attachment)

Dennis Miller, REAC Program Manager (w/o attachment)

0232\DFR\083002

FPXRF Results

The highest Zn result via FPXRF was determined on sample 29 at 28,000 milligrams per kilogram (mg/kg), the highest As result via FPXRF was determined on sample 44 at 1,600 mg/kg, and the highest Pb result via FPXRF was determined on sample 9-2 at 18,000mg/kg. Duplicate sample XRF cups were prepared whenever possible. FPXRF analyses of duplicate cups were determined and both results are presented in Table 1. In most cases the RPD of the duplicates were less than 50 percent (%). In the cases where the RPD was above 50% the XRF sample was reanalyzed (sample suffix "RR"), the outlier results was omitted, and a new RPD determined. In all cases this resulted in a RPD for duplicate FPXRF analyses below 50%.

Reference samples NIST SRM #2720, 2711, and R33, with known values of Zn, As, and Pb, were analyzed via FPXRF. The RPD was determined comparing only data sets above the FPXRF MQLs. In all cases the RPDs for NIST SRM samples were below 50%.

All Spectrace 9000 FPXRF daily instrument checkout criteria were met during the sampling period, as per SOP #1713. The comparison of FPXRF analysis of duplicate XRF cups and NIST SRM reference samples were acceptable during the sampling period (RPD< 50%).

TAL Metals Results

The highest Zn TAL metals result was determined on sample 29 at 59,000 mg/kg, the highest As TAL metals result was determined on sample 44 at 1,500 mg/kg, and the highest Pb TAL metals result was determined on sample 61 at 95,000 mg/kg. Three replicate samples were submitted for TAL metals analysis and the RPDs determined. In all case the RPDs were below 50 %. Table 2 summarizes the TAL metals confirmatory laboratory results.

FPXRF vs. TAL Metals Results

Table 3 summarizes the comparison of FPXRF field results and the corresponding TAL metals confirmatory laboratory results. The overall RPD for the target metal was less 50% with Zn at 24.8%, As at 49.8%, and Pb at 51%. Linear regression analysis of the FPXRF field analysis vs. TAL metals confirmatory laboratory analysis yielded the following R^2 values: Zn = 0.96, As = 0.71, Pb (all values above MDL) = 0.45, and Pb (all values above MDL, below 10,000 mg/kg) = 0.71. The regression analysis shows that the FPXRF field results were sufficiently validated ($R^2 > 0.7$) against the confirmatory laboratory analysis for Zn and As at all the ranges tested, and for Pb only at concentrations below 10,000 mg/kg. As such, all Zn and As FPXRF data can be considered QA-2 level data and only Pb FPXRF data below 10,000 mg/kg can be considered QA-2 level data. FPXRF results above 10,000 mg/kg Pb are therefore suspect and remain QA-1 level data. Appendix A has the complied QA/QC data and linear regression results.

Table I
Qualified FPXRF Screening Results
Yankee Mine Site
Unita National Forest, Utah
August 2002

Sample	Date	Zinc (Zn)	Arsenic (As)	Lead (Pb)
SAMPI	2 October 2001	630	150 J	800
SAMPIDUP	2 October 2001	780	150 J	610
SAMP2	2 October 2001	5400	240 J	4300
SAMP2RR	2 October 2001	1900	U	4500
SAMP2DUP	2 October 2001	1900	U (500)	5000
SAMP3	2 October 2001	530	U (72)	2100
SAM3DUP1	2 October 2001	830	U (72)	2200
SAM3DUP2	2 October 2001	520	U (72)	2000
SAMP4	2 October 2001	4700	U (72)	8800
SAMP4DUP	2 October 2001	5400	U (72)	8300
SAMP5	2 October 2001	1800	U (72)	4200
SAMP5DUP	2 October 2001	1400	U (72)	3600
SAMP6	2 October 2001	890	U (72)	1700
SAMP6DUP	2 October 2001	1000	U (72)	1800
SAMP7	2 October 2001	2600	U (72)	7200
SAMP7DUP	2 October 2001	3000	U (72)	7200
SAMP8	2 October 2001	200 J	U (72)	88 J
SAMP8DUP	2 October 2001	200 J	U (72)	98 J
SAMP9	2 October 2001	420	150 J	480
SAMP9DUP	2 October 2001	240 J	140 J	470
SAMP10	2 October 2001	2800	U (160)	1600
SAM10DUP_	2 October 2001	2600	180 J	1400
SAMP11	2 October 2001	Ū	U (72)	1600
SAMIIDUP	2 October 2001	_110 J	U (72)	1900
SAMP12	2 October 2001	1800	540	520
SAMP13	2 October 2001	290 J	U (72)	1100
SAMP14	2 October 2001	990	220 J	870
SAMP15	2 October 2001	600	100 J	180
SAMP16	2 October 2001	680	150 J	320
SAMP17	2 October 2001	560	U (200)	2000
SAMP18	2 October 2001	7500	U (72)	6100
SAM18DUP	2 October 2001	6800	U (72)	5400
SAMP19	2 October 2001	160 J	U (110)	1100
SAMP20	2 October 2001	2100	U (72)	4600
SAMP21	3 October 2001	U	U (72)	260
SAMP22	3 October 2001	_330 J	130 J	1200
SBMP23	3 October 2001	1000	350	1700
SAM23DUP	3 October 2001	620	190 J	1700
SAMP24	3 October 2001	510	U (72)	2100
SAMP25	3 October 2001	1100	U (450)	4500
SAMP26	3 October 2001	730	U (72)	2600
SAMP51	3 October 2001	1100	U (72)	1100

Table 1 - Continued Qualified FPXRF Screening Results Yankee Mine Site Unita National Forest, Utah August 2002

Sample	Date	Zinc (Zn)	Arsenic (As)	Lead (Pb)
SAMSIDUP	3 October 2001	1100	U (72)	1100
SAMP52	3 October 2001	2100	U (72)	2600
SAM52DUP	3 October 2001	2300	U (72)	2400
SAMP53	3 October 2001	420	U (72)	230
SAM53DUP	3 October 2001	420	U (72)	280
SAMP27	3 October 2001	1200	U (72)	920
SAMP28	3 October 2001	8800	U (72)	13000 -
SAM28DUP	3 October 2001	11000	U (72)	15000
SAMP29	3 October 2001	28000	U (72)	16000
SAMP30	3 October 2001	1300	370	1600
SAMP32	3 October 2001	11000	U (72)	14000
SAMP33	3 October 2001	200 J	650	1200
SAMP34	3 October 2001	680	410	2200
SAMP35	3 October 2001	2800	600	4000
SAMP36	3 October 2001	3600	U (72)	3700
SAMP38	3 October 2001	250 J	U (140)	1400
SAM39DUP	3 October 2001	650	U (380)	3800
SAMP40	3 October 2001	2200	U (72)	2600
SAM41DUP	3 October 2001	160 J	130 J	670
SAMP31	3 October 2001	790	340	2700
SAMP37	3 October 2001	1700	350	2200
SAMP42	3 October 2001	120 J	U (72)	2400
SAMP43	3 October 2001	130 J	170 J	690
SAM43DUP	3 October 2001	310 J	130 J	870
SAMP44	3 October 2001	310 J	1600	410
SAMP45	3 October 2001	760	240 J	920
SAMP46	3 October 2001	170 J	U (72)	2400
SAMP47	3 October 2001	130 J	U (72)	2000
SAMM48	3 October 2001	600	U (480)	4800
SAMP49	3 October 2001	1600	U (290)	2900
SAMP50	3 October 2001	340	87 J	80 J
SAMP54	3 October 2001	1500	950	6800
SAM54DUP	3 October 2001	2700	810	5500
SAMP55	3 October 2001	3200	500	2200
SAMP56	3 October 2001	1700	U (910)	9100
SAM56DUP	3 October 2001	1200_	U (72)	7800
SAM56DRR	3 October 2001	1400	U (72)	7900
SAMP57	3 October 2001	1600	270	830
SAMP58	3 October 2001	960	100 J	480
SAMP59	3 October 2001	280 J	300	85 J
SAMP60	3 October 2001	3800	U (72)	1700
SAMP61	3 October 2001	14000_	U (72)	12000

Table 1 - Continued Qualified FPXRF Screening Results Yankee Mine Site Unita National Forest, Utah August 2002

Sample	Date	Zinc (Zn)	Arsenic (As)	Lead (Pb)
SAM61DUP	3 October 2001	15000	U (72)	13000
SAMP62	3 October 2001	820	320	120 J
SAMP63	3 October 2001	610	U (72)	4700
SAMP64	3 October 2001	13000	U (72)	16000
SAMP65	3 October 2001	300 J	U (72)	830
SAMP66	3 October 2001	320 J	710	530
SAMP67	3 October 2001	300 J	650	290
SAMP68	3 October 2001	1200	U (72)	4400
SAMP69	3 October 2001	870	470	160
SAMP70	3 October 2001	150 J	150 J	660
SAMP71	3 October 2001	820	U (72)	2200
SAMP72	3 October 2001	660	420	1200
SAMP73	3 October 2001	1500	U (72)	630
SAMP74	3 October 2001	1200	U (330)	3300
SAMP75	3 October 2001	2300	U (72)	9800
SAM75DUP	3 October 2001	3000	U (72)	11000
SAMP76	3 October 2001	1700	U (72)	4800
SAMP77	3 October 2001	450	U (72)	4900
SAMP78	3 October 2001	870	U (72)	4600
SAMP79	3 October 2001	1100	U (72)	4900
SAMP80	3 October 2001	630	U (72)	5600
SAMP81	3 October 2001	360	U (72)	3900
SAMP82	3 October 2001	670	U (72)	4000
SAMP83	3 October 2001	350	U (72)	1600
SAMP84	3 October 2001	190 J	U (270)	2700
SAMP85	3 October 2001	U	U (72)	3000
SAMP86	3 October 2001	140 J	120 J	900
SAM86DUP	3 October 2001	U	U (72)	900
SAMP87	4 October 2001	1100	U (72)	2100
SAMP88	4 October 2001	1500	380	430
SAMP89	4 October 2001	220 J	U (72)	U
SAMP90	4 October 2001	370	87 J	210
SAMP91	4 October 2001	1900	U (72)	4500
SAM91DUP	4 October 2001	2200	U (72)	4700
SAMP92	4 October 2001	1900	U (360)	3600
SAMP93	4 October 2001	3400	U (420)	4200
SAM93DUP	4 October 2001	2600	U (400)	4000
SAMP9 - 1	4 October 2001	470	810	3000
SAMP9 - 2	4 October 2001	6100	U (72)	18000
SAMP9 - 3	4 October 2001	5900	U (72)	14000
SAMP9 - 4	4 October 2001	430	740	360
SAMPI1 - I	4 October 2001	U	110 J	800

Table 1 - Continued Qualified FPXRF Screening Results Yankee Mine Site Unita National Forest, Utah August 2002

Sample	Date	Zinc (Zn)	Arsenic (As)	Lead (Pb)
SAMP11 - 2	4 October 2001	U	U (72)	1000
S94 - SURFACE	4 October 2001	21000	U (72)	4800
SAMP95	4 October 2001	140 J	U (72)	45 J
SAMP96	4 October 2001	650	U (72)	1900
SAMP96 - 1	4 October 2001	210 J	89 J	110 J
SAMP97	4 October 2001	2400	U (72)	8700
SAMP98	4 October 2001	560	U (72)	330

All results in parts per million by weight (mg/kg)

U = Results below method detection limit (MDL), As MDL value in parenthesis

J = Results above MDL but below quantitation limit (MQL)

Detection Limit	Zinc	Arsenic '	Lead
MDL	102	72	41
MQL	339	240	135

All detection limits are in parts per million by weight (mg/kg)

Arsenic = Arsenic MDL is 72 mg/kg, or 1/10th of corresponding Lead sample result, whichever is higher, when the Pb:As ratio is 5:1 or greater, Arsenic MDL value in parenthesis.

Table 2
Target Analyte List (TAL) Metal Confirmatory Laboratory Results
Yankee Mine Site
Unita National Forest, Utah
August 2002

	ARSENIC	LEAD	ZINC
Location	TAL	TAL	TAL
2*	650	11,000	3,400
4	430	16,000	14,000
7	490	20,000	5,600
10	120	2,600	4,700
10 (DUP)	170	2,300	4,700
12	360	560	1,500
14	290	880	1,600
18	680	20,000	11,000
20	140	4,800	1,500
23	300	1,700	1,300
23 (DUP)	300	1,800	920
25	400	3,800	1,100
28	370	32,000	23,000
29	280	21,000	59,000
32	270	17,000	21,000
35	1,200	5,200	4,400
44	1,500	340	160
50	43	86	240
54	820	21,000	6,700
54 (DUP)	820	25,000	5,100
59	230	95	190
61	560	95,000	29,000
64	520	23,000	20,000
75	160	30,000	5,700
86	81	930	46
97	300	11,000	2,600
9-2'	660	41,000	13,000
9-4'	560	310	360

All concentrations in parts per million, milligram per kilogram (mg/kg) ND = Not Detected at concentrations above the detection limit.

J = Estimated Value

Table 3
Comparison of XRF and TAL Metal Concentrations
Yankee Mine Site
Unita National Forest, Utah
August 2002

1	AR	SENIC	L	EAD	Z	INC
Location	TAL	XRF	TAL	XRF	TAL	XRF
2*	650	160	11,000	4,400	3,400	3,700
4	430	ND	16,000	8,800	14,000	4,700
7	490	NĎ	20,000	7,200	5,600	2,600
10	120	ND (160)	2,600	1,600	4,700	2,800
10 (DUP)	170	180	2,300	1,400	4,700	2,600
12	360	540	560	520	1,500	1,800
14	290	220 .	880	870	1,600	990
18	680	ND	20,000	6,100	11,000	7,500
20	140	ND	4,800	4,600	1,500	2,100
23	300	350	1,700	1,700	1,300	1,000
23 (DUP)	300	190	1,800	1,700	920	620
25	400	ND (450)	3,800	4,500	1,100	1,100
28	370	ND	32,000	13,000	23,000	8,800
29	280	ND	21,000	16,000	59,000	28,000
32	270	ND	17,000	14,000	21,000	11,000
35	1,200	600	5,200	4,000	4,400	2,800
4 4	1,500	1,600	340	410	160	310 J
50	43	87 J	86	80 J	240	340
54	820	950	21,000	6,800	6,700	1,500
54 (DUP)	820	810	25,000	5,500	5,100	2,700
59	230	300	95	85 J	190	280J
61	560	ND	95,000	12,000	29,000	14,000
64	520	ND	23,000	16,000	20,000	13,000
75	160	ND	30,000	9,800	5,700	2,300
86	81	120 J	930	900	46	140 J
97	300	ND	11,000	8,700	2,600	2,400
9-2'	660	ND	41,000	18,000	13,000	6,100
9-4'	560	740	310	360	360	430

All concentrations in parts per million, milligram per kilogram (mg/kg) ND = Not Detected at concentrations above the detection limit. $J = Estimated\ Value$

Appendix A
FPXRF MDL, MQL, and QA/QC Data
FPXRF Final Trip Report
Yankee Mine Site
Unita National Forest, Utah
August 2002

Appendix A FPXRF MDL and MQL Raw Data

oct 2001	Spectrace U	Init Q-011					
FPXRF met	thod detection lim	it (MDL), metho	od quantitation	limit (MQL) raw data		
yankee mine	e wa# 0-232	· _					
60 sec each	source, NIST SRI	M #2709_					
	millign	ams/kilogram (n	ng/kg)				
date	Zinc (Zn) Arsenic (As) Lead (Pb)					Rounded	i (mg/kg)
oct 2	160	91	-19			MDL	MQL
oct 2	145	14	8.4		Zn	102	339
oct 2	181	37	-22		As	72	240
oct 2	170	33_	4	-	РЪ	41	135
oct 2	151	34	5				
oct 2	85	63_	-22				
oct 2	183	81_	4.8				
oct 2	162	43	3.7				
oct 3	119	38	-16				
oct 3	191	7.6	6.7				
oct 3	92	60_	-15				
oct 3	181	24	-25				
oct 3	142	70	4.1				
oct 3	118	46	10				
oct 3	137	78	-9.6				
oct 3	188	55	-19				
oct 4	121	45	-25				
oct 4	127	77_	1.9				
oct 4	145	14_	0.02				
oct 4	198	62	5.3				
oct 4	97	72	17				
n=21							
std dev	33.886786	23.968074	13.512079				
MDL	101.66036	71.904222	40.536237				
MQL	338.86786	239,68074	135.12079	-			

Appendix A - Continued QA/QC and Linear Regression Data Data Yankee Mine Site

Unita National Forest Utah July 2002

TAL														
TAL														
TAL	ARSENIC (A	\s)		<u> </u>		L	Ŀ	1		<u></u>	'		L	
	XRF	RPD	CALC-As	 	Regression Out	nud: Ac > *	#DI							
					Predicaziou Cat	<u>⊬ui. ∧s > R</u>		<u> </u>						 -
43	87	67.69		Constant	L		. 0				!			
81	120	38.81	118 32	Std Err of	Y Fet		235,1327							
						<u> </u>	-		_					1
650	160	120.99		R Squared		ļ	0.703408	<u></u>						
170	180	5.71	177.47	No. of Obs	ervations	İ	14	į l			l i			
300	190	44.90		Degrees o			13							 -
				Degrees o	rieedoni									
290	220	27.45	216.91			<u> </u>		<u> </u>						
230	300	26.42	295.79	X Coefficie	ent(s)	0.985968						. — —		1
300	350	15.38				0.098531								
				Std Err of	ÇOBI.	0.090331								+ -
360	540	40.00	532.42			L								
1,200	600	66.67	591.58											
	740				VOC							-		-
560		27.69	729.62		x = XRF	ļ								
820	810	1.23	798.63		v = TAL		ļ	l .	١ .					ł
820	950	14.69	936.67							_		_		
												-		+
1,500	1,600	6.45	1577.55		Mean RPD =	24.82	(>MQL)							<u> </u>
			1											1
	LEAD (Pb	· · · · · ·												_
										 -				}
TAL	XRF	RPD	CALC-Pb		Regression Out	put: all Pb	> MDL			Regressio	n O <u>utput: P</u>	b > MDL <	<u>10,</u> 000	
86	80	7.23	191 97	Constant			1 0		Constant			oi		
					V E			 i		V E4		_		
95	85	11,11		Std Err of		<u> </u>	14519.07	<u> </u>	Std Err of		ļ .	4821.296	<u>!</u>	
310	360	14.93	863,85	R Squared	l '	L	0.451722	L l	R Squared		l ¹	0.741167	L	
340	410	18.67		No. of Obs		-	28		No. of Obs			22		
								 -						
560	520	7.41		Degrees o	r readom		27		Degrees o	r rreedom		21		
880	870	1,14	2087.64											1
930	900	3.28		X Coefficie	nt(e)	2.39959	1		X Coefficie	unt(e)	2.382619			
							•							+
2,300	1,400	48.65		Std Err of	Coef.	<u>0.337659</u>	4.12E-17	<u> </u>	Std Err of	Coef	0.214773		<u></u>	
2,600	1,600	47.62	3639.34											
1,800	1,700	5.71			· · · · · · · · · · · · · · · · · · ·	_	-							
-			4079.30		·		<u> </u>							1
1,700	1,700	0.00	4079.30		x = XRF		<u> </u>				<u> </u>		<u></u>	
5,200	4,000	26.09	9598.36		y = TAL							_		1
11,000	4,400	85.71		i	<u> </u>	·····							-	+
			10558.20	ļ .		<u> </u>	<u> </u>						<u> </u>	-
3,800	4,500	16.87	10798.16		Mean RPD =	49.76	(>MQL)							
4,600	4,600	4.26	11038.11						_	_				
25,000	5,500	127.87												
			13197.75				_							<u> </u>
20,000	6,100	106.51	14637.50											1
21,000	6,800	102.16	16317.21										• •	
														
20,000	7,200	94,12	17277.05								1			
11,000	8,700	23.35	20876.43			·						i		
16,000	8,800	58.06	21116.39	_		_								
							<u> </u>							
30,000	9,800	101.51	23515.98				<u> </u>							ł
95.000	12,000	155,14	28795.06											1
32,000	13,000													+
		84.44	31194.67				_					_		
17,000	14,000	19.35	33594.26					1			1			1
21,000	16,000	27.03	38393.44											
23,000				-		·· · ·								
	16,000	35.90	38393.44				L							
41,000	18,000	77.97	43192.62											
														-
	ZINC (Zn)													
TAL	XRF	RPD			Regression Out	nutrall 7n	NACI							
46	140		ICALC-Zn I											
			CALC-Zn		Tregression car	put. dii En					_			<u> </u>
		101,08	281.96	Constant		put. all Ell	0			_				
190	280	101,08 38,30	281.96 563.91	Constant Std Err of	Y Est	put. dii zii								
		101,08	281.96 563.91	Constant	Y Est	pot. dii 211	0 2484.346							
190 160	280 310	101,08 38,30 63,83	281.96 563.91 624.33	Constant Std Err of R Squared	Y Est	put. dii 211	0 2484,346 0.961401							
190 160 240	280 310 340	101,08 38.30 63.83 34.48	281.96 563.91 624.33 684.75	Constant Std Err of R Squared No. of Obs	Y Est ervations	put. all Ell	0 2484,346 0.961401 28							
190 160 240 360	280 310 340 430	101,08 38.30 63.83 34.48 17.72	281.96 563.91 624.33 684.75 866.01	Constant Std Err of ' R Squared No. of Obs Degrees of	Y Est ervations	Jul. dii Eli	0 2484,346 0.961401							
190 160 240	280 310 340	101,08 38.30 63.83 34.48 17.72	281.96 563.91 624.33 684.75 866.01	Constant Std Err of ' R Squared No. of Obs Degrees of	Y Est ervations	Jul. dii Eli	0 2484,346 0.961401 28							
190 160 240 360 920	280 310 340 430 620	101,08 38,30 63,83 34,48 17,72 38,96	281.96 563.91 624.33 684.75 866.01 1248.66	Constant Std Err of R Squared No. of Obs Degrees of	Y Est ervations f Freedom		0 2484,346 0.961401 28							
190 160 240 360 920 1,600	280 310 340 430 620 990	101,08 38.30 63.83 34.48 17,72 38.96 47.10	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie	Y Est ervations f Freedom	2,013972	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300	280 310 340 430 620 990 1,000	101,08 38,30 63,83 34,48 17,72 38,96 47,10 26,09	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97	Constant Std Err of R Squared No. of Obs Degrees of	Y Est ervations f Freedom		0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600	280 310 340 430 620 990	101,08 38,30 63,83 34,48 17,72 38,96 47,10 26,09	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie	Y Est ervations f Freedom	2,013972	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100	280 310 340 430 620 990 1,000	101,08 38,30 63,83 34,48 17,72 38,96 47,10 26,09 0.00	281,96 563,91 624,33 684,75 866,01 1248,66 1993,83 2013,97 2215,37	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie	Y Est ervations f Freedom	2,013972	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700	280 310 340 430 620 990 1,000 1,100 1,500	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37	Constant Std Err of R Squared No. of Obs Degraes of X Coefficie Std Err of	Y Est ervations f Freedom int(s) Coef.	2,013972	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500	280 310 340 430 620 990 1,000 1,100 1,500	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15	Constant Std Err of R Squared No. of Obs Degraes of X Coefficie Std Err of	Y Est ervations f Freedom	2,013972	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500	280 310 340 430 620 990 1,000 1,100 1,500	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15	Constant Std Err of R Squared No. of Obs Degraes of X Coefficie Std Err of	Y Est ervations f Freedom int(s) Coef. x = XRF	2,013972	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500	280 310 340 430 620 990 1.000 1.100 1,500 1,800 2,100	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34	Constant Std Err of R Squared No. of Obs Degraes of X Coefficie Std Err of	Y Est ervations f Freedom int(s) Coef.	2,013972	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700	280 310 340 430 620 990 1.000 1.100 1.500 1,800 2,100 2,300	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 2215.37 3020.96 3625.15 4229.34 4632.14	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,300 2,400	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom int(s) Coef. x = XRF	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,300 2,400	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600 4,700	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,300 2,400 2,600	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.14 4632.14 4833.53 5236.33	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 1,500 2,600 4,700 5,600	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600 4,700	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,300 2,400 2,600	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.14 4632.14 4833.53 5236.33	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 2,600 2,600 4,700 5,600 5,100	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,700	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1.600 1.300 1.100 6.700 1.500 1.500 5.700 2.600 4.700 5.600 5.100	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,300 2,400 2,600 2,600 2,700 2,800	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 2,600 2,600 4,700 5,600 5,100	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,700 2,800 2,800 2,800	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5437.72 5639.12	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1.600 1.300 1.100 6.700 1.500 1.500 5.700 2.600 5.600 5.100 4.700 4.400	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,700 2,800 2,800 2,800	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 6.00 57.53 73.17 61.54 50.67	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5437.72 5639.12	Constant Std Err of R Squared No. of Obs Degrees o X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1.600 1.300 1.100 6.700 1.500 1.500 5.700 2.600 4.700 5.500 5.100 4.700 4.400 3.400	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,600 2,800 2,800 2,800 2,800 2,800 3,700	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5236.33 5236.33 5236.33 5236.33	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1.600 1.300 1.100 6.700 1.500 1.500 5.700 2.600 4.700 5.500 4.700 4.400 3.400	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,300 2,400 2,600 2,600 2,600 2,700 2,800 2,800 3,700 4,700	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5236.37.72 5639.12 7451.70 9465.67	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1.600 1.300 1.100 6.700 1.500 1.500 5.700 2.600 4.700 5.500 5.100 4.700 4.400 3.400	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,600 2,800 2,800 2,800 2,800 2,800 3,700	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5236.33 5236.33 5236.33 5236.33	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 5,700 2,600 4,700 5,600 5,100 4,700 4,700 4,400 4,400 3,400 14,000	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,600 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 3,700 4,700 6,100	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5437.72 5639.12 5639.12 7451.70 9465.67	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 5,700 2,600 4,700 5,600 5,100 4,700 4,400 3,400 14,000 13,000	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,700 2,800 2,800 3,700 4,700 6,100 7,500	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25 37.84	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.36 4632.14 4632.14 4833.53 5236.33 5236.33 5437.72 5639.12 7451.70 9465.67 12285.23 15104.79	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 5,700 2,600 4,700 5,600 5,100 4,700 4,700 4,400 4,400 3,400 14,000	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,600 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 2,800 3,700 4,700 6,100	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5437.72 5639.12 5639.12 7451.70 9465.67	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600 4,700 4,700 4,400 3,400 14,000 13,000 13,000	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,400 2,600 2,600 2,600 2,700 2,800 2,800 3,700 4,700 6,100 7,500 8,800	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25 37.84 89.31	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5437.72 5639.12 7451.70 9465.67 12285.23 15104.79 17722.95	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600 4,700 4,700 4,400 3,400 14,000 13,000 13,000 11,000 21,000	280 310 340 430 620 990 1,000 1,100 1,500 2,100 2,300 2,400 2,600 2,600 2,700 2,800 2,800 2,800 3,700 4,700 6,100 7,500 8,800 11,000	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25 37.84 89.31 62.50	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5437.72 5639.12 7451.70 9465.67 12285.23 15104.79 17722.95 22153.69	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600 4,700 4,470 4,470 4,400 3,400 13,000 11,000 11,000 20,000	280 310 340 430 620 990 1,000 1,100 1,500 2,300 2,400 2,600 2,600 2,600 2,700 2,800 3,700 4,700 6,100 7,500 8,800 11,000 13,000	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25 37.84 89.31 62.50 42.42	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5236.33 5437.72 5639.12 7451.70 9465.67 12285.23 15104.79 17722.95 22153.69 26181.64	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600 4,700 4,470 4,470 4,400 3,400 13,000 11,000 11,000 20,000	280 310 340 430 620 990 1,000 1,100 1,500 2,300 2,400 2,600 2,600 2,600 2,700 2,800 3,700 4,700 6,100 7,500 8,800 11,000 13,000	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25 37.84 89.31 62.50 42.42	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5236.33 5437.72 5639.12 7451.70 9465.67 12285.23 15104.79 17722.95 22153.69 26181.64	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom ont(s) Coef. x = XRF y = TAL	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 1,500 5,700 2,600 4,700 4,470 4,470 4,400 3,400 13,000 11,000 13,000 11,000 20,000 20,000 20,000	280 310 340 430 620 990 1,000 1,100 1,500 1,800 2,100 2,300 2,400 2,600 2,600 2,700 2,800 3,700 4,700 6,100 7,500 8,800 11,000 13,000 14,000	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25 37.84 89.31 62.50 42.42 69.77	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5437.72 5639.12 7451.70 9465.67 12285.23 15104.79 17722.95 22153.69 26181.64 28195.61	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom int(s) Coef. x = XRF y = TAL Mean RPD =	2.013972 0.063344	0 2484.346 0.961401 28 27							
190 160 240 360 920 1,600 1,300 1,100 6,700 1,500 1,500 5,700 2,600 4,700 4,470 4,470 4,400 3,400 13,000 11,000 11,000 20,000	280 310 340 430 620 990 1,000 1,100 1,500 2,300 2,400 2,600 2,600 2,600 2,700 2,800 3,700 4,700 6,100 7,500 8,800 11,000 13,000	101.08 38.30 63.83 34.48 17.72 38.96 47.10 26.09 0.00 126.83 18.18 33.33 85.00 8.00 57.53 73.17 61.54 50.67 44.44 8.45 99.47 72.25 37.84 89.31 62.50 42.42	281.96 563.91 624.33 684.75 866.01 1248.66 1993.83 2013.97 2215.37 3020.96 3625.15 4229.34 4632.14 4833.53 5236.33 5236.33 5236.33 5236.33 5437.72 5639.12 7451.70 9465.67 12285.23 15104.79 17722.95 22153.69 26181.64	Constant Std Err of R Squared No. of Obs Degrees of X Coefficie Std Err of	Y Est ervations f Freedom int(s) Coef. x = XRF y = TAL Mean RPD =	2.013972 0.063344	0 2484.346 0.961401 28 27							

Appendix B
Daily FPXRF Field Screening Results
FPXRF Final Trip Report
Yankee Mine Site
Unita National Forest, Utah
August 2002

Site Name: Orem, Utah

Units: ppm

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Sample	Date	Zn	Zn		As	As	
Code	Run	Raw	Qual		Raw	Qual	
			=========		======================================	**======	:====
	Time: 2-0	290.00	290	J	59.00	ND	-
SAMP1	2-OCT-2001	626.06	630	-	152.40	150	J
SAMP1DUP	2-OCT-2001	783.95	780	-	153.05	150	J
SAMP2	2-OCT-2001	5386.45	5400	-	243.29	240	J
SAMP2RR	2-OCT-2001	1880.66	1900	-	43.41	ND	-
SAMP2DUP	2-OCT-2001	1946.48	1900	-	142.44	140	J
SAMP3	2-OCT-2001	532.26	530	-	-12.32	ND	-
SAM3DUP1	2-OCT-2001	830.23	830	-	-44.75	ND	-
SAM3DUP2	2-OCT-2001	515. 9 9	520	-	58.14	ND	-
SAMP4	2-QCT-2001	4674.58	4700	-	-554.56	ND	-
SAMP4DUP	2-OCT-2001	5388.93	5400	-	-218.76	ИD	-
SAMP5	2-OCT-2001	1794.15	1800	-	-438.54	ND	-
SAMP5DUP	2-OCT-2001	1431.51	1400	-	- 277.28	ND	-
SAMP6	2-OCT-2001	885.43	890	-	~10.72	ND	-
SAMP6DUP	2-OCT-2001	995.38	1000	-	~76.57	NÐ	-
SAMP7	2-OCT-2001	2557.64	2600	-	-431.65	ND	-
SAMP7DUP	2-OCT-2001	2990.95	3000	-	~364.65	ИD	-
SAMP8	2-OCT-2001	202.45	200	J	6.00	ND	-
SAMP8DUP	2-OCT-2001	198.02	200	J	-15.40	ND	-
SAMP9	2-OCT-2001	415. 9 8	420	-	153.46	150	J
SAMP9DUP	2-OCT-2001	241.71	240	J	142.54	140	J
SAMP10	2-OCT-2001	2759.85	2800	-	90.50	91	J
SAM10DUP	2-QCT-2001	2558.87	2600	-	177.76	180	J
SAMP11	2-OCT-2001	91.75	ND	-	0.00	ND	-
SAM11DUP	2-OCT-2001	105.14	110	J	-71.84	ND	-
SAMP12	2-OCT-2001	1830.37	1800	-	543.04	540	-
SAMP13	2-OCT-2001	289.55	290	J	58.71	ND	_

Application: SOILS with U, Th, Ag Q011 07-08-1992

Minimum Detection Limit (MDL) = 93 79
Minimum Quantitation Limit (MQL) = 311 264

ND = below MDL

J = above MDL, below MQL

Site Name: Orem, Utah

Units: ppm

=========				=====		=======	====
Sample	Date	2n	Zn		As	As	
Code	Run	Raw =======	Qual		Raw =======	Qual	_====
SAMP14	2-OCT-2001	991.44	990	_	215.18	220	J
SAMP15	2-OCT-2001	603.18	600	-	103.92	100	J
SAMP16	2-OCT-2001	679.13	680	-	146.72	150	J
SAMP17	2-OCT-2001	559.98	560	_	88.21	88	J
SAMP18	2-OCT-2001	7479.15	7500	-	-14.03	ND	_
SAM18DUP	2-OCT-2001	6776.94	6800	-	-218.82	ND	_
SAMP19	2-OCT-2001	164.29	160	J	100.58	100	J
SAMP20	2-OCT-2001	2063.95	2100	_	-98.78	ND	-

Application: SOILS with U, Th, Ag Q011 07-08-1992

Minimum Detection Limit (MDL) = 93 79
Minimum Quantitation Limit (MQL) = 311 264

ND = below MDL

J = above MDL, below MQL

Site Name: Orem, Utah

Units: ppm

C02=======		3========		***********************
Sample	Date	Pb	₽b	
Code	Run	Raw	Oual	
===========				
SAMP14	2-OCT-2001	869.53	870	-
SAMP15	2-OCT-2001	178.21	180	-
SAMP16	2-OCT-2001	316.90	320	-
SAMP17	2-OCT-2001	1984.16	2000	-
SAMP18	2-OCT-2001	6141.07	6100	-
SAM18DUP	2-OCT-2001	5441.38	5400	-
SAMP19	2-OCT-2001	1143.28	1100	-
SAMP20	2-OCT-2001	4632.24	4600	-

7malication.COTIC with D mb 3m 0011 07 00 1000

Application: SOILS with U, Th, Ag Q011 07-08-1992

Pb

Minimum Detection Limit (MDL) = 41
Minimum Quantitation Limit (MQL) = 137

ND = below MDL

J = above MDL, below MQL

Site Name: American Fork Canyon, Orem, Utah

Units: ppm

=======================================	*****	=======	-==	=======	==:		
Sample Code	Date Run	Zn Qual		As Qual		Pb Qual	
=======================================	a=======		===		==:		
SAMP21	3-OCT-2001	ND	-	ND	_	260	-
SAMP22	3-OCT-2001	330	J	130	J	1200	-
SBMP23	3-OCT-2001	1000	_	350	-	1700	-
SAM23DUP	3-OCT-2001	620	-	190	J	1700	-
SAMP24	3-OCT-2001	510	-	ИĎ	-	2100	-
SAMP25	3-OCT-2001	1100	_	390	-	4500	-
SAMP26	3-OCT-2001	730	-	ND	-	2600	-
SAMP51	3-OCT-2001	1100	-	ND	-	1100	-
SAM51DUP	3-OCT-2001	1100	-	ND	_	1100	•
SAMP52	3-OCT-2001	2100	-	ND	_	2600	-
SAM52DUP	3-OCT-2001	2300	-	ND	_	2400	-
SAMP53	3-OCT-2001	420	-	ND	_	230	-
SAM53DUP	3-OCT-2001	420	-	ND	_	280	-
SAMP27	3-OCT-2001	1200	-	NĎ	-	920	-
SAMP28	3-OCT-2001	8800	_	ND	-	13000	-
SAM28DUP	3-OCT-2001	11000	-	ND	-	15000	-
SAMP29	3-OCT-2001	28000	-	ИD	-	16000	-
SAMP30	3-OCT-2001	1300	_	370	-	1600	-
SAMP32	3-OCT-2001	11000	-	ND	-	14000	-
SAMP33	3-OCT-2001	200	J	650	-	1200	-
SAMP34	3-OCT-2001	680	-	410	-	2200	-
SAMP35	3-OCT-2001	2800	-	600	-	4000	-
SAMP36	3-OCT-2001	3600	-	ND	_	3700	-
SAMP38	3-0CT-2001	250	J		J	1400	-
SAM39DUP	3-OCT-2001	650	-	160	J	3800	-
SAMP40	3-OCT-2001	2200	-	ND	-	2600	-
SAM41DUP	3-OCT-2001	160	J	130	J	670	-

Application: SOILS with U, Th, Ag Q011 07-08-1992

Minimum Detection Limit (MDL) = 102 72 41
Minimum Quantitation Limit (MQL) = 339 240 135
ND = below MDL

J = above MDL, below MQL

Site Name: American Fork Canyon, Orem, Utah

Units: ppm

		_	===		===		
Sample	Date	Zn		As		Pb	
Code	Run	Qual		Qual		Qual	
SAMP31	3-0CT-2001	790		340	===:	2700	_
SAMP31 SAMP37	3-0CT-2001	1700	_	350		2200	-
SAMP42	3-0CT-2001	120		ND	_	2400	
SAMP42 SAMP43	3-0CT-2001	130	-	170	_	690	_
SAM43DUP	3-0CT-2001	310		130		870	
SAMP44	3-0C1-2001 3-0CT-2001	310		1600	_		~
		760			-	410	-
SAMP45	3-OCT-2001		-	240	_	920	-
SAMP46	3-OCT-2001	170	J		-	2400	-
SAMP47	3-OCT-2001	130	Ų	ND	_	2000	-
SAMM48	3-OCT-2001	600	-	110		4800	-
SAMP49	3-OCT-2001	1600	-	230	_	2900	<u>-</u>
SAMP50	3-OCT-2001	340	-	87	J		J
SAMP54	3-OCT-2001	1500	-	950		6800	-
SAM54DUP	3-OCT-2001	2700	-	810	-	5500	~
SAMP55	3-OCT-2001	3200	-	500	-	2200	-
SAMP56	3-OCT-2001	1700	_	350	_	9100	-
SAM56DUP	3-OCT-2001	1200	-	ND	-	7800	-
SAM56DRR	3-0CT-2001	1400	-	ND	-	7900	-
SAMP57	3-OCT-2001	1600	-	270	-	830	-
SAMP58	3-OCT-2001	960	-	100	J	480	-
SAMP59	3-OCT-2001	280	J	300	_	85	J
SAMP60	3-0CT-2001	3800	_	ND	-	1700	-
SAMP61	3-OCT-2001	14000	_	ЙD	_	12000	_
SAM61DUP	3-OCT-2001	15000	_	ND	_	13000	_
SAMP62	3-OCT-2001	820	_	320	_	120	J
SAMP63	3-0CT-2001	610	_	ND	_	4700	- -
SAMP64	3-OCT-2001	13000	_	ND	_	16000	_
SAMP65	3-OCT-2001	300	J		_	830	_
SAMP66	3-OCT-2001	320	J	710	_	530	_
SAMP67	3-0CT-2001	300	J		_	290	_
==========							==========

Application: SOILS with U, Th, Ag Q011 07-08-1992

	2n	As	Pb
Minimum Detection Limit (MDL) =	102	72	41
Minimum Quantitation Limit (MQL)=	339	240	135
ND = below MDL			

J = above MDL, below MQL

Site Name: American Fork Canyon, Orem, Utah

Units: ppm

==========		========	==:		==:	========	
Sample	Date	Zn		As		Pb	
Code	Run	Qual		Qual		Qual	
=======================================			==:		==:		
SAMP68	3-OCT-2001	1200	_	ND	_	4400	-
SAMP69	3-OCT-2001	870	-	470	-	160	-
SAMP70	3-OCT-2001	150	J	150	J	660	-
SAMP71	3-OCT-2001	820	-	ND	-	2200	-
SAMP72	3-OCT-2001	660	-	420	-	1200	-
SAMP73	3-OCT-2001	1500	-	ND	-	630	-
SAMP74	3-OCT-2001	1200	-	150	J	3300	-
SAMP75	3-OCT-2001	2300	-	ND	٠-	9800	-
SAM75DUP	3-OCT-2001	3000	-	NÐ	-	11000	-
SAMP76	3-OCT-2001	1700	-	ND	-	4800	-
SAMP77	3-OCT-2001	450	-	ND	-	4900	-
SAMP78	3-OCT-2001	870	-	ND	-	4600	-
SAMP79	3-OCT-2001	1100	-	ND	-	4900	-
SAMP80	3-OCT-2001	630	_	ND	-	5600	-
SAMP81	3-OCT-2001	360	-	ND	-	3900	-
SAMP82	3-OCT-2001	670	-	ND	-	4000	-
SAMP83	3-OCT-2001	350	-	ND	-	1600	-
SAMP84	3-OCT-2001	190	J	160	J	2700	-
SAMP85	3-OCT-2001	ND	-	ND	-	3000	-
SAMP86	3-OCT-2001	140	J	120	J	900	-
SAM86DUP	3-0CT-2001	ND	-	ND	-	900	-

Application: SOILS with U, Th, Ag Q011 07-08-1992

ND = below MDLJ = above MDL, below MQL

Site Name: American Fork Canyon, Orem, Utah

Units: ppm

=========	==========	=======			
Sample	Date	Zn	As	Pb	
Code	Run	Qual	Qual	Qual	
========		========			
SAMP87	4-OCT-2001	1100	- ND	- 2100	-
SAMP88	4-OCT-2001	1500	- 380	- 430	-
SAMP89	4-QCT-2001	220	J ND	- ND	_
SAMP90	4-OCT-2001	370	- 87	J 210	_
SAMP91	4-OCT-2001	1900	- ND	- 4500	-
SAM91DUP	4-OCT-2001	2200	– ИD	- 4700	-
SAMP92	4-OCT-2001	1900	- 240	- 3600	-
SAMP93	4-OCT-2001	3400	- 270	- 4200	_
SAM93DUP	4-OCT-2001	2600	- 230	J 4000	-
SAMP9-1	4-OCT-2001	470	- 810	- 3000	-
SAMP9-2	4-OCT-2001	6100	- ND	- 18000	-
SAMP9-3	4-OCT-2001	5900	- ND	- 14000	_
SAMP9-4	4-OCT-2001	430	- 740	- 360	-
SAMP11-1	4-OCT-2001	ND	- 110	J 800	-
SAMP11-2	4-OCT-2001	ND	- ND	- 1000	_
S94-SUR	4-0CT-2001	21000	- ND	- 4800	_
SAMP95	4-OCT-2001	140	J ND	- 45	J
SAMP96	4-OCT-2001	650	- ND	- 1900	_
SAMP96-1	4-OCT-2001	210	J 89	J 110	J
SAMP97	4-OCT-2001	2400	- ND	- 8700	-
SAMP98	4-OCT-2001	560	- ND	- 330	-

Application: SOILS with U, Th, Ag Q011 07-08-1992

Minimum Detection Limit (MDL) = 102 72 41 Minimum Quantitation Limit (MQL) = 339 240 135 ND = below MDL

J = above MDL, below MQL

Site Name: Orem, Utah

Units: ppm

		========	
Date	₽b	Pb	
Run	Raw	Qual	
		===========	
Time: 2-0	1148.00	1100	-
2-OCT-2001	797.98	800	-
2-OCT-2001	613.45	610	-
2-OCT-2001	4292.76	4300	-
2-OCT-2001	4451.86	4500	-
2-OCT-2001	5008.29	5000	-
2-0CT-2001	2089.69	2100	-
2-OCT-2001	2216.74	2200	-
2-OCT-2001	2029.25	2000	-
2-OCT-2001	8843.44	8800	-
2-OCT-2001	8257.95	8300	-
2-OCT-2001	4188.49	4200	_
2-OCT-2001	3633.67	3600	_
2-OCT-2001	1693.30	1700	-
2-OCT-2001	1753.00	1800	-
2-OCT-2001	7247.29	7200	-
2-OCT-2001	7245.51	7200	-
2-OCT-2001	87.51	88	J
2-OCT-2001	98.04	98	J
2-OCT-2001	475.46	480	-
2-OCT-2001	467.26	470	-
2-OCT-2001	1565.22	1600	-
2-OCT-2001	1350.51	1400	-
2-OCT-2001	1631.43	1600	-
2-OCT-2001	1897.09	1900	_
2-OCT-2001	521.23	520	-
2-OCT-2001	1147.67	1100	-
	Date Run Time: 2-0 2-OCT-2001	Date Run Raw Time: 2-0 1148.00 2-OCT-2001 797.98 2-OCT-2001 4292.76 2-OCT-2001 4451.86 2-OCT-2001 5008.29 2-OCT-2001 2089.69 2-OCT-2001 2029.25 2-OCT-2001 8257.95 2-OCT-2001 8257.95 2-OCT-2001 3633.67 2-OCT-2001 3633.67 2-OCT-2001 1693.30 2-OCT-2001 7247.29 2-OCT-2001 7245.51 2-OCT-2001 98.04 2-OCT-2001 475.46 2-OCT-2001 467.26 2-OCT-2001 1565.22 2-OCT-2001 1631.43 2-OCT-2001 1897.09 2-OCT-2001 1897.09 2-OCT-2001 521.23	Run Raw Qual Time: 2-0 1148.00 1100 2-OCT-2001 797.98 800 2-OCT-2001 613.45 610 2-OCT-2001 4292.76 4300 2-OCT-2001 4451.86 4500 2-OCT-2001 5008.29 5000 2-OCT-2001 2089.69 2100 2-OCT-2001 2089.69 2100 2-OCT-2001 2029.25 2000 2-OCT-2001 2029.25 2000 2-OCT-2001 8843.44 8800 2-OCT-2001 8257.95 8300 2-OCT-2001 4188.49 4200 2-OCT-2001 3633.67 3600 2-OCT-2001 1693.30 1700 2-OCT-2001 1753.00 1800 2-OCT-2001 7247.29 7200 2-OCT-2001 7247.29 7200 2-OCT-2001 87.51 88 2-OCT-2001 475.46 480 2-OCT-2001 475.46 480 2-OCT-2001 1565.22 1600 2-OCT-2001 1565.22 1600 2-OCT-2001 1565.22 1600 2-OCT-2001 1631.43 1600 2-OCT-2001 1897.09 1900 2-OCT-2001 1897.09 1900 2-OCT-2001 521.23 520

Application: SOILS with U, Th, Ag Q011 07-08-1992

Pb

Minimum Detection Limit (MDL) = 41 Minimum Quantitation Limit (MQL) = 137

ND = below MDL

J = above MDL, below MQL

Appendix C
Copies of Field Logbook Entries
FPXRF Final Trip Report
Yankee Mine Site
Unita National Forest, Utah
August 2002

电影公理电影,然后还是这个时间的正式,因此就是这种是一个人的情况,但是是一个人的情况,但是是一个人的人的人们是一个人的人们是一个人的人们是一个人们的人们们是一个人们们

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3	San 1/0	1452	2760	9/	1565	Locature #10
7	SAM ID AUD	16 39	3	170	1251	dep Locaco
+	Sandy	11.00	92	-30	// 3/	hecipon & P
İ	50mg /1 010	1662	76	7	1897	du Locall
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-(,	samp 19	/807	164	101	11.43	Locaha W
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	#2709	1818	181	37	-22	QC -SRM
ļ	全2710	1827	16346	56	4920	QC -SRM
į	42711	1826	572	8.2	1045	OC - SRM
	# R33	1831	120	281	270	OC-SRM
,	4 SAND	1855	153	-8.0	1.19	9c-500
ٳٞ	£2709	1840	328	60	14	QC-SRY
	\$ 2709	1844	37/	89	172	QC-5PM
-	#2709	20,48	170	3.3	4.0	OC -SAM
[12709	2054	151	34	50	OC-SEM
į	162709	2057	85	63	- 27	OC-SRH
	# 2709	2/03	183	81	4.8	OC-JRM
	# 2709	2108	162	4 √3	3.7	OC-SRM
Ì	# £33	2112	0,9	234	351	OC-SRM
	#27/0	21 17	5830	200	4670	
5	2711	2/22	321	37	1020	DA - CRA
	#Smo	2128	-7.6	-15	1.9	QC-BIAN
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34 PROJECT	YANKER Mine	WAF 232	Notebook No	33
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	DATE 16/3/01		SPECTRACE SERIAL NO	
	SITE Youkee Mind		WA# WA# 2	each
} 	Source: Cd 109 Pb La	2,57 - 12.65)	Source: Fe55 S Ka	Range KeV(10.49 - 10.59) KeV(12.56 - 12.66)
	IRON Ka RESC		TY CHECK (Cd 109, II → 2739	RON PURE)
* Ichcio	Iron at maximum peak height (MP 1/2 MF		Counts (MPH ≥ 1000 at 6 counts	•
	left (low energy) side, 1/2 MPF /368 counts at 6.262 /639 counts at 6.278	子KeV, 1013	right (high energy) side /29 counts at counts at	6.5371 KeV (0.
	Calculated FWH	M= () 2 (10)	KeV (≤ 0.300)	4
: 10	D Fail: Counts ≤ 1/2 Ml	PH at 6.25 KeV	Cd 109 Intensity Check Fe 0,990239 Mn 0,0012909	Criteria $(\ge 0.95 \text{ and } \le 1.05)$ $\cancel{y}(\le \pm 0.006)$
# SAND	Pass □ Fail: Counts ≤ 1/2 Mi	PH at 6.55 Kev	Co - 0,000/0788	E (≤ ± 0.006)
		BLANK SAM		_
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. BO	- (7° - ~1 A	lements Cr (z=24) and high	ner are within ± 5 std. deviation	
	Comments 56 8	$\frac{3.20}{\text{As}}$	1,6/7-/6	-11.41141
-	NOTE: All acquisition times	≥60 seconds each source	, All checks with standard S Initials _	Max.
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Jankee Mone.	Site		, ,		
4. 0 4. 0		opm (ug/	9-)		
Sample Frue		#3	1420	Comments	
	25.4 650	[0]	1420 3820		
Sam 3900 /209 Samp 40 /215		-120	2618		
Sand duy 1220	27.50	131	1068		
Sam aug 120	156		068		
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	039.Res 4	a u	u vra	get ex	
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Samp 31 1258	790	340	2728	withy organics	
SAmp 37 /253	1737	3 Sv	2167	Location 437	
samp 42 1257	121	-128	2412	4 4 42	
Samp 43 1308	126	173	692	4 4 43	
Samp 4300P 1366	312	129	872	dup Locat 43	. ,
Samp 44 1321	310	1562	409	Locomon# 40	1
Samp 45 1324	764	235	915	4 45	-
Samo 46 /3321	/72	- 39	2376	4 46	,
Samo 47 /337	/31	-151_	2045	1 1 47 1 48	
samo 48 /341 #2709 - 1346	142	// 0	4760		
	1670	70 230	2940	QC-SRM	
2000	339	8-2	-29	Location # 49	•
Samo 54 1400	74/2	950	6800	30	
Sam 54 dup 1405	2730	810	5460	1 SIL 11 SIL	,
Samo 55 14 10	3200	500	9727	I no me	•
Sam 0 56 14 14	1750	350	9080		
Sam 0 56 dup 14 LO	1120	- 44	7790		
SANS6 D RR 1425	135a	-67		rerund dup Lic #5	56
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Appendix C

Appendix C
Final Analytical Report
Yankee Mine Site
Unita National Forest, Utah
August 2002

Lockheed Martin Technology Services
Environmental Services REAC
2890 Woodbridge Avenue Building 209 Annex
Edison, NJ 08837-3679
Telephone 732-321-4200 Facsimile 732-494-4021



DATE:

17 December 2001

TO:

R. Singhvi

EPA/ERTC

FROM:

D. Miller

Analytical Section Leader

Leader D M

SUBJECT:

DOCUMENT TRANSMITTAL UNDER WORK ASSIGNMENT # 0-232

Attached please find the following document prepared under this work assignment:

Yankee Mine - Analytical Report

Central File WA # 0-232

A. Humphrey

S. Grossman

J. Soroka

(w/attachment)

Work Assignment Manager (w/attachment)

Task Leader (w/attachment)

Data Validation and Report Writing Group Leader (w/o attachment)

ANALYTICAL REPORT

Prepared by LOCKHEED MARTIN, Inc.

> Yankee Mine **New York**

December 2001

EPA Work Assignment No. 0-232 LOCKHEED MARTIN Work Order R1A00232 EPA Contract No. 68-C99-223

> Submitted to A. Humphrey **EPA-ERTC**

S. Grossman

Analysis by:

REAC

Task Leader

D. Miller

Prepared by:

G. Karustis

Analytical Section Leader

Reviewed by:

J. Soroka

Program Manager

Table of Contents

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Appendices will be furnished on request.

Introduction

REAC in response to WA 0-232, provided analytical support for environmental samples collected from Yankee Mine, located in New York as described in the following table. The support also included QA/QC, data review, and preparation of an analytical report containing a summary of the analytical methods, the results, and the QA/QC results.

The samples were treated with procedures consistent with those specified in SOP #1008.

COC#	Number of Samples	Sampling Date	Date Received	Matrix	Analysis	Laboratory	Data Package
00358	16	10/3/01	10/9/01	Water	Metals	REAC	K 371
00536	9	10/1/01	10/9/01	Soil	Metals	REAC	K 369
00536	10	10/2/01	10/9/01	Soil	Metals	REAC	K 369
00537	7	10/3/01	10/9/01	Soil	Metals	REAC	K 368
00537	1	10/2/01	10/9/01	Soil	Metals	REAC	K 368
00537	1	10/4/01	10/9/01	Soil	Metals	REAC	K 368

The samples were received in XRF cups. Percent moistures were not determined due to the small sample size,

Case Narrative

The data in this report have been validated to two significant figures. Any other representation of the data is the responsibility of the user.

Metals in Water Package K 371

The acceptable QC limits for the percent recovery were exceeded for lead in sample A 04460 MSD (59%). The results of the lead analysis for the associated samples A 04454 through A 04460 should be regarded as estimated.

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Metals in Soil Package K 369

The acceptable QC limits for the percent recoveries of antimony, barium and selenium were exceeded as tabulated below.

	B 03607 MS	B 03607 MSD
Antimony Barium Selenium	22 Acceptable 127	1 297 Acceptable
	B 03616 MS	B 03616 MSD
Antimony Selenium	Acceptable 21	37 13

The data are affected as follows:

The results of the antimony analysis for samples B 03601 through B 03616, B 03619 and B 03620 should be regarded as estimated.

The results of the selenium analysis for samples B 03601 through B 03603, B 03608, B 03609, B 03612 throuthB 03616 and B 03619 should be regarded as estimated.

The results of the barium analysis for samples B 03601 through B 03607 should be regarded as estimated.

Metals in Soil Package K 368

The acceptable QC limits for the percent recoveries of antimony and zinc were exceeded as tabulated below.

	<u>B 03625 MS</u>	B 03625 MSD
Antimony	23	27
Zinc	632	Acceptable

The data are affected as follows:

The results of the antimony analysis for samples B 03622 through B 03625 should be regarded as estimated.

The results of the barium analysis for samples B 03621 through B 03628 should be regarded as estimated.

Summary of Abbreviations

AΑ Atomic Absorption В The analyte was found in the blank BFB Bromofluorobenzene С Centigrade Continued cont. D (Surrogate Table) this value is from a diluted sample and was not calculated (Result Table) this result was obtained from a diluted sample Dioxin and/or PCDD and PCDF denotes Polychlorinated Dibenzo-p-dioxins and Polychlorinated Dibenzofurans CLP Contract Laboratory Protocol COC Chain of Custody CONC Concentration CRDL Contract Required Detection Limit Contract Required Quantitation Limit CRQL DFTPP Decafluorotriphenylphosphine **Detection Limit** DL The value is greater than the highest linear standard and is estimated Ε **EMPC** Estimated maximum possible concentration ICAP Inductively Coupled Argon Plasma ISTD Internal Standard The value is below the method detection limit and is estimated LCS Laboratory Control Sample LCSD Laboratory Control Sample Duplicate MDL **Method Detection Limit** MΙ Matrix Interference MS (BS) Matrix Spike (Blank Spike) Matrix Spike Duplicate (Blank Spike Duplicate) MSD (BSD) MW Molecular Weight NA either Not Applicable or Not Available NC Not Calculated NR Not Requested NS Not Spiked % D Percent Difference % REC Percent Recovery PPB Parts per billion **PPBV** Parts per billion by volume Parts per million by volume **PPMV** PQL **Practical Quantitation Limit** Quality Assurance/Quality Control QA/QC QL Quantitation Limit

RPD Relative Percent Difference
RSD Relative Standard Deviation
SIM Selected Ion Monitoring

TCLP Toxic Characteristics Leaching Procedure

U Denotes not detected

W Weathered analyte; Aroclor pattern displays a degradation of earlier eluting peaks m^3 kilogram cubic meter kg microgram μg gram liter pg picogram : g milliliter mg milligram mL ng nanogram

μL microliter

denotes a value that exceeds the acceptable QC limit

Abbreviations that are specific to a particular table are explained in footnotes on

that table

Revision 7/26/01

ANALYTICAL REPORT

Prepared by LOCKHEED MARTIN, inc.

> Yankee Mine **New York**

December 2001

EPA Work Assignment No. 0-232 LOCKHEED MARTIN Work Order R1A00232 EPA Contract No. 68-C99-223

> Submitted to A. Humphrey **EPA-ERTC**

S. Grossman

Task Leader

D. Miller

Analytical Section Leader

S. Clapp Program Manager

Date

Date

Analysis by:

REAC

Prepared by:

G. Karustis

Reviewed by:

J. Soroka

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00537	11	10/2/01	10/9/01	Soil	Metals	REAC	K 368
00537	1	10/4/01	10/9/01	Soil	Metals	REAC	K 368

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Metals in Water Package K 371

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The acceptable QC limits for the percent recoveries of antimony, barium and selenium were exceeded as tabulated below.

	B 03607 MS	<u>B 03607 MSD</u>
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	B 03616 MS	B 03616 MSD
Antimony Selenium	Acceptable 21,	37 13

The data are affected as follows:

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Metals in Soil Package K 368

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The results of the barium analysis for samples B 03621 through B 03628 should be regarded as estimated.

	Summary of Abbreviations	
AA	Atomic Absorption	
В	The analyte was found in the blank	
BFB	Bromofluorobenzene	
C	Centigrade	
cont.	Continued	
D	(Surrogate Table) this value is from a diluted sample and was not ca	alculated
_	(Result Table) this result was obtained from a diluted sample	
Dioxin and/or		
	CDF denotes Polychlorinated Dibenzo-p-dioxins and Polychlorinated Dil	benzofurans
CLP	Contract Laboratory Protocol	
COC	Chain of Custody	
CONC	Concentration	
CRDL	Contract Required Detection Limit	
CRQL	Contract Required Quantitation Limit	
DFTPP	Decafluorotriphenylphosphine	
DL	Detection Limit	
E	The value is greater than the highest linear standard and is estimate	ed
EMPC	Estimated maximum possible concentration	
ICAP	Inductively Coupled Argon Plasma	
ISTD	Internal Standard	
J	The value is below the method detection limit and is estimated	
LCS	Laboratory Control Sample	
LCSD	Laboratory Control Sample Duplicate	
MDL	Method Detection Limit	
MI	Matrix Interference	
MS (BS)	Matrix Spike (Blank Spike)	
MSĎ (BSD)	Matrix Spike Duplicate (Blank Spike Duplicate)	
MW	Molecular Weight	
NA	either Not Applicable or Not Available	
NC	Not Calculated	
NR	Not Requested	
NS	Not Spiked	
% D	Percent Difference	
% REC	Percent Recovery	
PPB	Parts per billion	
PPBV	Parts per billion by volume	
PPMV	Parts per million by volume	
PQL	Practical Quantitation Limit	
QA/QC	Quality Assurance/Quality Control	
QL	Quantitation Limit	
RPD	Relative Percent Difference	
RSD	Relative Standard Deviation	
SIM	Selected Ion Monitoring	
TCLP	Toxic Characteristics Leaching Procedure	
U	Denotes not detected	
w	Weathered analyte: Aroclor pattern displays a degradation of earlier	
.m ₃		nicrogram
L _.		oicogram
mμ		nanogram
1/1	microliter	

 $_{\bullet}^{\mu\mathsf{L}}$ microliter

denotes a value that exceeds the acceptable QC limit

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> Submitted to A. Humphrey **EPA-ERTC**

S. Grossman

Task Leader

Date

Analysis by:

REAC

D. Miller

Prepared by:

G. Karustis

Analytical Section Leader

Program Manager

Reviewed by:

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Barium	Acceptable	297
Selenium	127	Acceptable
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Antimony	Acceptable	37
Selenium	21,	13

The data are affected as follows:

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Summary of Abbreviations

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^~	Albinic Absorbtion

B The analyte was found in the blank

BFB Bromofluorobenzene

C Centigrade cont. Continued

D (Surrogate Table) this value is from a diluted sample and was not calculated

(Result Table) this result was obtained from a diluted sample

Dioxin and/or

PCDD and PCDF denotes Polychlorinated Dibenzo-p-dioxins and Polychlorinated Dibenzofurans

CLP Contract Laboratory Protocol

COC Chain of Custody CONC Concentration

CRDL Contract Required Detection Limit
CRQL Contract Required Quantitation Limit
DFTPP Decafluorotriphenylphosphine

DL Detection Limit

E The value is greater than the highest linear standard and is estimated

EMPC Estimated maximum possible concentration

ICAP Inductively Coupled Argon Plasma

ISTD Internal Standard

J The value is below the method detection limit and is estimated

LCS Laboratory Control Sample

LCSD Laboratory Control Sample Duplicate

MDL Method Detection Limit
MI Matrix Interference
MS (BS) Matrix Spike (Blank Spike)

MSD (BSD) Matrix Spike Duplicate (Blank Spike Duplicate)

MW Molecular Weight

NA either Not Applicable or Not Available

NC Not Calculated
NR Not Requested
NS Not Spiked
% D Percent Difference
% REC Percent Recovery
PPB Parts per billion

PPBV Parts per billion by volume
PPMV Parts per million by volume
PQL Practical Quantitation Limit
QA/QC Quality Assurance/Quality Control

QL Quantitation Limit

RPD Relative Percent Difference
RSD Relative Standard Deviation
SIM Selected Ion Monitoring

TCLP Toxic Characteristics Leaching Procedure

U Denotes not detected

W Weathered analyte; Aroclor pattern displays a degradation of earlier eluting peaks m^3 cubic meter kg kilogram microgram μg gram liter g pg picogram milligram milliliter mg nanogram mL ng

μL microliter

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REAC in response to WA 0-232, provided analytical support for environmental samples collected from Yankee Mine, located in New York as described in the following table. The support also included QA/QC, data review, and preparation of an analytical report containing a summary of the analytical methods, the results, and the QA/QC results.

The samples were treated with procedures consistent with those specified in SOP #1008.

COC#	Number of Samples	Sampling Date	Date Received	Matrix	Analysis	Laboratory	Data Package
00358	16	10/3/01	10/9/01	Water	Metals	REAC	K 371
00536	9	10/1/01	10/9/01	Soil	Metals	REAC	K 369
00536	10	10/2/01	10/9/01	Soil	Metals	REAC	K 369
00537	7	10/3/01	10/9/01	Soil	Metals	REAC	K 368
00537	1	10/2/01	10/9/01	Soil	Metals	REAC	K 368
00537	1	10/4/01	10/9/01	Soil	Metals	REAC	K 368

The samples were received in XRF cups. Percent moistures were not determined due to the small sample size.

Case Narrative

The data in this report have been validated to two significant figures. Any other representation of the data is the responsibility of the user.

Metals in Water Package K 371

The acceptable QC limits for the percent recovery were exceeded for lead in sample A 04460 MSD (59%). The results of the lead analysis for the associated samples A 04454 through A 04460 should be regarded as estimated.

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Metals in Soil Package K 369

The acceptable QC limits for the percent recoveries of antimony, barium and selenium were exceeded as tabulated below.

	<u>B 03607 MS</u>	<u>B 03607 MSD</u>
Antimony Barium Selenium	22 Acceptable 127	1 297 Acceptable
	<u>B 03616 MS</u>	B 03616 MSD
Antimony Selenium	Acceptable 21	37 13

The data are affected as follows:

The results of the antimony analysis for samples B 03601 through B 03616, B 03619 and B 03620 should be regarded as estimated.

The results of the selenium analysis for samples B 03601 through B 03603, B 03608, B 03609, B 03612 throuthB 03616 and B 03619 should be regarded as estimated.

The results of the barium analysis for samples B 03601 through B 03607 should be regarded as estimated.

Metals in Soil Package K 368

The acceptable QC limits for the percent recoveries of antimony and zinc were exceeded as tabulated below.

	B 03625 MS	B 03625 MSD
Antimony	23	27
Zinc	632	Acceptable

The data are affected as follows:

The results of the antimony analysis for samples B 03622 through B 03625 should be regarded as estimated.

The results of the barium analysis for samples B 03621 through B 03628 should be regarded as estimated.

	Summary of Abbro	eviations		
AA	Atomic Absorption			
В	The analyte was found in the blan	ık		
BFB	Bromofluorobenzene			
С	Centigrade			
cont,	Continued			
D	(Surrogate Table) this value is from (Result Table) this result was obtain			t calculated
Dioxin and/or	, ,			
PCDD and PCI	DF denotes Polychlorinated Dibenzo	o-p-dioxins and	Polychlorinated	Dibenzofurans
CLP	Contract Laboratory Protocol	- •	,	
COC	Chain of Custody			
CONC	Concentration			
CRDL	Contract Required Detection Limit	t		
CRQL	Contract Required Quantitation Li			
DFTPP	Decafluorotriphenylphosphine	*****		_
DL	Detection Limit			
E	The value is greater than the high	est linear stand	lard and is actim	satad
EMPC	Estimated maximum possible con	centration	iaio and is estin	ialeo
ICAP	Inductively Coupled Argon Plasma			
ISTD	Internal Standard	a		
J	The value is below the method de	tection limit on	d in actimated	
Ľcs	Laboratory Control Sample	RECTION MINE AND	ıs eştimateti	
	Laboratory Control Sample Duplic	anto.		
LCSD	Method Detection Limit	ate		
MDL				
MI	Matrix Interference			
MS (BS)	Matrix Spike (Blank Spike)	I BP4-A		
MSD (BSD)	Matrix Spike Duplicate (Blank Spi	ke Duplicate)		
MW	Molecular Weight	_6.1.		
NA NG	either Not Applicable or Not Availa	able		
NC	Not Calculated			
NR NR	Not Requested			
NS	Not Spiked			
% D	Percent Difference			
% REC	Percent Recovery			
PPB	Parts per billion			
PPBV	Parts per billion by volume			
PPMV	Parts per million by volume			
PQL	Practical Quantitation Limit			
QA/QC	Quality Assurance/Quality Control	1		
QL	Quantitation Limit			
RPD	Relative Percent Difference			
RSD	Relative Standard Deviation			
SIM	Selected Ion Monitoring			
TCLP	Toxic Characteristics Leaching Pr	rocedure		
U	Denotes not detected			
W	Weathered analyte; Arocior patter		egradation of ear	
m³		kilogram	μ g	microgram
L		gram	pg	picogram
mL		milligram	ng	nanogram
1	microliter	_		-

 $\mu\mathsf{L}$

denotes a value that exceeds the acceptable QC limit

Abbreviations that are specific to a particular table are explained in footnotes on

that table

microliter

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ANALYTICAL REPORT

Prepared by LOCKHEED MARTIN, inc.

> Yankee Mine **New York**

December 2001

EPA Work Assignment No. 0-232 LOCKHEED MARTIN Work Order R1A00232 EPA Contract No. 68-C99-223

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Metals in Water Package K 371

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The data are affected as follows:

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The results of the barium analysis for samples B 03621 through B 03628 should be regarded as estimated.

	Summary of Abbreviations
AA	Atomic Absorption
В	The analyte was found in the blank
BFB	Bromofluorobenzene
С	Centigrade
cont.	Continued
D	(Surrogate Table) this value is from a diluted sample and was not calculated
	(Result Table) this result was obtained from a diluted sample
Dioxin and/or	
PCDD and PC	DF denotes Polychlorinated Dibenzo-p-dioxins and Polychlorinated Dibenzofurans
CLP	Contract Laboratory Protocol
COC	Chain of Custody
CONC	Concentration
CRDL	Contract Required Detection Limit
CRQL	Contract Required Quantitation Limit
DFTPP	Decafluorotriphenylphosphine -
ĎΓ	Detection Limit
E	The value is greater than the highest linear standard and is estimated
EMPC	Estimated maximum possible concentration
ICAP	Inductively Coupled Argon Plasma Internal Standard
ISTD J	The value is below the method detection limit and is estimated
rcs	Laboratory Control Sample
LCSD	Laboratory Control Sample Duplicate
MDL	Method Detection Limit
MI	Matrix Interference
MS (BS)	Matrix Spike (Blank Spike)
MSD (BSD)	Matrix Spike Duplicate (Blank Spike Duplicate)
MW	Molecular Weight
NA	either Not Applicable or Not Available
NC	Not Calculated
NR	Not Requested
NS	Not Spiked
% D	Percent Difference
% REC	Percent Recovery
PPB	Parts per billion
PPBV	Parts per billion by volume
PPMV	Parts per million by volume
PQL	Practical Quantitation Limit
QA/QC	Quality Assurance/Quality Control
QL	Quantitation Limit
RPD	Relative Percent Difference
RSD	Relative Standard Deviation
SIM	Selected Ion Monitoring
TCLP	Toxic Characteristics Leaching Procedure
Ü	Denotes not detected
m ₃	Weathered analyte; Aroclor pattern displays a degradation of earlier eluting peaks
	cubic meter kg kilogram μ g microgram liter g gram pg picogram
Ն mL	The state of the s
πιι μL	milliliter mg milligram ng nanogram microliter
# *	denotes a value that exceeds the acceptable QC limit
	Abbreviations that are specific to a particular table are explained in footnotes on
	that table
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Analytical Procedure for Metals in Water

Sample Preparation

A representative 45 mL aliquot of each sample was mixed with 5.0-mL concentrated nitric acid, placed in an acid rinsed Teflon container, capped with a Teflon lined cap, and digested according to SW-846, method 3015 in a CEM MDS-2100 microwave oven, which was programmed to bring the samples to 160 +/- 4°C in 10 minutes (first stage) and slowly to 165-170°C in the second 10 minutes (second stage). After digestion, the samples were allowed to cool to room temperature and were transferred to acid cleaned bottles. The samples were analyzed for all metals, except mercury, by US EPA SW-846, method 7000 Atomic Absorption (AA) or method 6010 Inductively Coupled Argon Plasma (ICAP) procedures.

A 100 mL aliquot of each sample was transferred to a 300-mL BOD bottle and prepared according to SW-846, method 7470. The samples were heated for 2 hours on a hot plate at 95° C, cooled to room temperature and reduced with hydroxylamine hydrochloride (NH₂OH:HCl). Mercury was then analyzed separately on a Leeman Labs PS200II AA Spectrometer.

A reagent blank and a blank spike sample were carried through the sample preparation procedure for each analytical batch of samples processed. One matrix spike (MS) and one matrix spike duplicate (MSD) sample were also processed for each analytical batch or every 10 samples.

Analysis and Calculations

The AA, ICAP and Leeman Labs PS200II instruments were calibrated and operated according to SW-846, method 7000/7470/6010 and the manufacturer's operating instructions. After calibration, initial calibration verification (ICV), initial calibration blank (ICB), and QC check standards were run to verify proper calibration. The continuing calibration verification (CCV) and continuing calibration blank (CCB) standards were run after every 10 samples to verify proper operation during sample analysis.

The metal concentration in solution, in micrograms per liter (mg/L), was read directly from the readout system of the instrument. ICAP and mercury results were taken directly from instrument readouts. The ICAP results were corrected for digestion volume (45-mL sample + 5-mL nitric acid) prior to instrument read-out; AA read-outs (excluding mercury) were externally corrected for digestion volume (1.1111 * AA read-out).

For samples that required dilution to fall within the instrument calibration range:

mg/L metal in sample = A [(C+B) / C]

where:

A = direct read-out (ICAP and mercury)

A = corrected read-out (AA)

B = acid blank matrix used for dilution, mL

C = sample aliquot, mL

Results of the analyses are listed in Table 1.1.

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Analytical Procedure for Metals in Soil

Sample Preparation

A representative 1-2 g (wet weight) sample, weighed to 0.01 g accuracy, was mixed with 10-mL 1:1 nitric acid, placed in a 50-mL polypropylene digestion cup and digested in nitric acid and hydrogen peroxide according to SW-846, Method 3050 B on a Hot Block digestion system. The final reflux was either nitric acid or hydrochloric acid depending on the metals to be determined. After digestion, the samples were allowed to cool to room temperature, transferred to 100 mL volumetric flasks and diluted to volume with ASTM Type II water. The samples were analyzed for all metals, except mercury, by USEPA SW-846, Method 7000 (Atomic absorption) or Method 6010 (Inductively Coupled Argon Plasma-ICAP) procedures.

A representative 0.25-0.8 g (wet weight) sample was transferred to a 300-mL BOD bottle and prepared according to SW-846, Method 7471. The sample was heated for 1/2 hour on a hot plate at 95° C, cooled to room temperature, and reduced with hydroxylamine hydrochloride (NH₂OH:HCl). Mercury was then analyzed separately on a Leeman Labs PS200II AA Spectrometer.

A separate sample was used to determine total solids.

A reagent blank and a blank spike sample were carried through the sample preparation procedure for each batch of samples processed. One matrix spike (MS) and one matrix spike duplicate (MSD) were analyzed for each batch or for every ten samples.

Analysis and Calculations

The AA, ICAP and Leeman Labs PS200II instruments were calibrated and operated according to SW-846, Method 7000/7471/6010 and the manufacturers operating instructions. After calibration, initial calibration verification (ICV), initial calibration blank (ICB) and quality control check standards were run to verify proper calibration. The continuing calibration verification (CCV) and continuing calibration blank (CCB) were run after every ten samples to assure proper operation during sample analysis.

The metal concentration in solution, in micrograms per liter (µg/L) was taken from the read-out system of the Atomic Absorption instrument. The results were converted to milligrams per kilogram (mg/kg) by correcting the reading for the sample weight and percent solids. The ICAP results (mg/kg) were corrected for sample weight prior to instrument read-out; the instrument read-out was then corrected for percent solids.

Final concentrations, based on wet weight are given by:

mg metal/kg sample = [(AxV)/W]xDFxCF

where:

A = Instrument read-out (µg/L, AA; mg/kg, ICAP)

V = final volume of processed sample (mL, AA; 1.00 ICAP)

W = weight of sample (g, AA; 1.00 ICAP)

DF = Dilution Factor (1.00 for no dilution)

CF = conversion factor (0.001, AA; 1.00, ICAP)

For samples that required dilution to be within the instrument calibration range, DF is given by:

DF = (C+B)/C

where:

B = acid blank matrix used for dilution (mL)

C = sample blank aliquot (mL)

Final concentrations, based on dry weight, are given by:

mg/kg(dry) = [mg/kg (wet)x100] /S

where

S = percent solids

The results are listed in Table 1.2.

Revision date: 8/17/2000

Table 1.1 Results of the Analysis for Metals in Water WA # 0-232 Yankee Mine Site

Client ID Location		Method La		A04 SW		A044 SW-01		A044 SW-		A04 SW		A04 SW	
Parameter	Analysis Method	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL μg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L
Aluminum	ICAP	U	50	480	50	480	50	U	50	340	50	260	50
Antimony	AA-Fur	U	2.2	U	2.2	U	2.2	Ū	2.2	Ü	2.2	Ũ	2.2
Arsenic	AA-Fur	Ū	2.2	Ū	2.2	Ū	2.2	Ū	2.2	2.8	2.2	4.0	2.2
Barium	ICAP	Ų	5.0	23	5.0	23	5.0	61	5.0	79	5.0	24	5.0
Beryllium	ICAP	Ü	2.0	Ų	2.0	U	2.0	U	2.0	U	2.0	U	2.0
Cadmium	ICAP	Ü	5.0	U	5.0	Ü	5.0	U	5.0	U	5.0	U	5.0
Calcium	1CAP	Ų	100	20000	100	20000	100	37000	100	48000	100	64000	100
Chromium	ICAP	U	5.0	U	5.0	U	5.0	U	5.0	U	5.0	U	5.0
Cobalt	ICAP	U	10	Ų	10	U	10	U	10	U	10	Ü	10
Copper	ICAP	U	10	44	10	45	10	Ų	10	U	10	Ū	10
Iron	ICAP	U	25	37	25	39	25	U	25	400	25	590	25
Lead	AA-Fur	U	2.2	U	2.2	U	2.2	U	2.2	11	2.2	12	2.2
Magnesium	ICAP	U	500	7700	500	7800	500	19000	500	25000	500	33000	500
Manganese	ICAP	U	5.0	16	5.0	17	5.0	Ų	5.0	25	5.0	29	5.0
Mercury	Cold Vapor	Ų	0.20	U	0.20	U	0.20	U	0.20	U	0.20	Ú	0.20
Nickel	ICAP .	U	10	U	10	U	10	U	10	U	10	U	10
Potassium	ICAP	υ	2000	U	2000	U	2000	Ų	2000	υ	2000	Ū	2000
Selenium	AA-Fur	U	2.2	U	2.2	U	2.2	Ų	2.2	U	2.2	Ù	2.2
Silver	ICAP	Ū	5	U	5.0	Ü	5.0	Ú	5.0	Ū	5.0	Ŭ	5.0
Sodium	ICAP	Ü	500	770	500	750	500	750	500	820	500	900	500
Thallium	AA-Fur	U	2.2	U	2.2	Ų	2.2	Ų	2.2	U	2.2	Ü	2.2
Vanadium	ICAP	Ū	10	U	10	Ú	10	Ų	10	Ū	10	Ŭ	10
Zinc	ICAP	Ū	10	44	10	44	10	26	10	21	10	34	10

Table 1.1 (cont.) Results of the Analysis for Metals in Water WA # 0-232 Yankee Mine Site

Client ID Location		A04 SW		A04- SW	. • .	A044 SW		A044 SW-		A04 SW		A04 SW	
Parameter	Analysis Method	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL μg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L
Aluminum	ICAP	100	50	υ	50	140	50	3300	50	1600	50	560	50
Antimony	AA-Fur	2.9	2.2	U	2.2	4.6	2.2	46	2.2	13	2.2	6.5	2.2
Arsenic	AA-Fur	83	2.2	U	2.2	U	2.2	920	11	98	2.2	49	2.2
Barium	ICAP	15	5.0	78	5.0	26	5.0	67	5.0	68	5.0	36	5.0
Beryllium	ICAP	U	2.0	U	2.0	U	2.0	2.3	2.0	U	2.0	U	2.0
Cadmium	ICAP	Ú	5.0	U	5.0	U	5.0	U	5.0	7.4	5.0	Ü	5.0
Calcium	ICAP	23000	100	44000	100	24000	100	25000	100	30000	100	29000	100
Chromium	ICAP	U	5.0	U	5.0	U	5.0	U	5.0	Ü	5.0	Ü	5.0
Cobalt	ICAP	U	10	U	10	Ų	10	Ų	10	Ú	10	Ú	10
Copper	ICAP	U	10	Ų	10	12	10	110	10	140	10	37	10
Iron	ICAP	8200	25	77	25	250	25	120000	25	14000	25	5700	25
Lead	AA-Fur	2.9	2.2	5.2	2.2	6.0	2.2	230	2.2	180	2.2	57	2.2
Magnesium	ICAP	8800	500	15000	500	9600	500	9700	500	12000	500	11000	500
Manganese	ICAP	140	5.0	5.2	5.0	19	5.0	200	5.0	110	5.0	79	5.0
Mercury	Cold Vapor	Ų	0.20	Ų	0.20	Ų	0.20	U	0.20	0.28	0.20	U	0.20
Nickel	ICAP `	12	10	Ų	10	U	10	16	10	12	10	Ú	10
Potassium	ICAP	U	2000	U	2000	Ų	2000	U	2000	2500	2000	Ú	2000
Selenium	AA-Fur	U	2.2	Ú	2.2								
Silver	ICAP	Ų	5.0	U	5.0	U	5.0	Ų	5.0	U	5.0	U	5.0
Sodium	ICAP	880	500	1400	500	770	500	900	500	990	500	870	500
Thallium	AA-Fur	U	2.2	U	2.2	Ų	2.2	86	2.2	U	2.2	U	2.2
Vanadium	ICAP	U	10	U	10	U	10	U	10	Ų	10	U	10
Zinc	ICAP	290	10	58	10	160	10	820	10	1000	10	320	10

Table 1.1 (cont.) Results of the Analysis for Metals in Water WA # 0-232 Yankee Mine Site

Client ID Location		A04 SW		A04- SW-		A044 SEE		A044 SEE			466 Blank
Parameter	Analysis Method	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L	Conc µg/L	MDL µg/L
Aluminum	ICAP	Ų	50	520	50	U	50	100	50	u	50
Antimony	AA-Fur	Ú	2.2	13	2.2	2.8	2.2	U	2.2	Ų	2.2
Arsenic	AA-Fur	U	2.2	63	2.2	8.4	2.2	U	2.2	Ú	2.2
Barium	ICAP	39	5.0	32	5.0	150	5.0	66	5.0	U	5.0
Beryllium	ICAP	U	2.0	Ų	2.0	U	2.0	U	2.0	υ	2.0
Cadmium	ICAP	Ų	5.0	U	5.0	U	5.0	5.0	5.0	υ	5.0
Calcium	ICAP	33000	100	27000	100	47000	100	44000	100	U	100
Chromium	ICAP	U	5.0	U	5.0	Ü	5.0	U	5.0	Ų	5.0
Cobalt	ICAP	U	10	Ų	10	Ų	10	Ų	10	U	10
Copper	ICAP	U	10	47	10	Ų	10	63	10	υ	10
Iron	ICAP	26	25	7300	25	8500	25	99	25	U	25
Lead	AA-Fur	Ų	2.2	130	2.2	U	2.2	11	2.2	Ų	2.2
Magnesium	ICAP	11000	500	11000	500	23000	500	22000	500	U	500
Manganese	ICAP	Ų	5.0	100	5.0	380	5.0	85	5.0	Ų	5.0
Mercury	Cold Vapor	U	0.20								
Nickel	ICAP	U	10	Ų	10	U	10	IJ	10	U	10
Potassium	ICAP	U	2000								
Selenium	AA-Fur	U	2.2	υ	2.2	U	2.2	U	2.2	Ų	2.2
Silver	ICAP	U	5.0	υ	5.0	U	5.0	U	5.0	U	5.0
Sodium	ICAP	900	500	850	500	1100	500	840	500	U	500
Thallium	AA-Fur	U	2.2								
Vanadium	ICAP	U	10	U	10	U	10	Ų	10	U	10
Zinc	ICAP	Ų	10	420	10	670	10	770	10	Ų	10

Client ID Location		Method La		B03		B03		B03	603	B03		B03 10 [
Parameter	Analysis Method	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg
Aluminum	ICAP	U	18	1300	17	1100	17	3600	17	4400	17	4300	17
Antimony	ICAP	U	6.0	330	5.8	210	5.7	370	5.8	69	5.8	83	5.8
Arsenic	ICAP	U	7.5	650	7.3	430	7.1	490	7.3	120	7.2	170	7.3
Barium	ICAP -	U	1.0	490	0.97	780	0.95	780	0.97	1000	0.96	840	0.97
Beryllium	ICAP	U	0.50	U	0.49	U	0.48	Ų	0.49	0.64	0.48	0.62	0.49
Cadmium	ICAP	U	0.50	25	0.49	110	0.48	43	0.49	36	0.48	35	0.49
Calcium	ICAP	Ü	50	330	49	330	48	350	49	4100	48	3900	49
Chromium	ICAP	Ú	0.50	1.2	0.49	1.0	0.48	5.8	0.49	6.2	0.48	6.3	0.49
Cobalt	ICAP	U	1.0	U	0.97	1.7	0.95	3.8	0.97	7.1	0.96	7.4	0.97
Copper	ICAP	Ų	1.0	500	0.97	79	0.95	280	0.97	340	0.96	220	0.97
iron	ICAP	U	10	28000	9.7	7900	9.5	23000	9.7	23000	9.6	22000	9.7
Lead	ICAP	υ	4.0	11000	3.9	16000	3.8	20000	3.9	2600	3.8	2300	3.9
Magnesium	ICAP	U	50	190	49	300	48	1200	49	3000	48	3000	49
Manganese	ICAP	U	1.0	19	0.97	170	0.95	110	0.97	440	0.96	440	0.97
Mercury	Cold Vapor	U	0.04	14	0.38	47	0.8	4.4	0.19	3.7	80.0	2.6	0.08
Nickel	ICAP	U	1.0	U	0.97	1.5	0.95	5.9	0.97	11	0.96	10	0.97
Potassium	ICAP	U	200	1800	190	670	190	1600	190	1400	190	1400	190
Selenium	AA-Fur	U	0.50	7.5	2.5	6.1	2.3	5.0	2.4	U	2.4	U	2.4
Silver	ICAP	U	0.50	33	0.49	62	0.48	46	0.49	13	0.48	13	0.49
Sodium	ICAP	U	50	94	49	U	48	U	49	υ	48	U	49
Thallium	AA-Fur	U	0.50	Ų	2.5	4.6	2.3	U	2.4	U	2.4	U	2.4
Vanadium	ICAP	Ū	2.0	4.4	1.9	2.0	1.9	8.1	1.9	8.6	1.9	8.3	1.9
Zinc	ICAP	U	2.0	3400	1.9	14000	9.5	5600	1.9	4700	1.9	4700	1.9

Client ID Location		B03		B03		B036 18		B03		B03		B03 23 D	611 DUP
Parameter	Analysis Method	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg
Aluminum	ICAP	10000	18	7200	18	540	17	6600	18	3400	17	3300	17
Antimony	ICAP	12	5.9	42	6.0	1400	5.7	85	5.9	77	5.8	87	5.8
Arsenic	ICAP	360	7.4	290	7.5	680	7.1	140	7.4	300	7.3	300	7.3
Barium	ICAP	310	0.98	150	1.0	47	0.94	230	0.99	740	0.97	600	0.97
Berytlium	ICAP	1.6	0.49	0.80	0.50	Ų	0.47	0.55	0.50	U	0.49	U	0.49
Cadmium	ICAP	8.5	0.49	13	0.50	76	0.47	11	0.50	11	0.49	6.7	0.49
Calcium	ICAP	8800	49	6700	50	980	47	7300	50	250	49	210	49
Chromium	ICAP	16	0.49	11	0.50	U	0.47	10	0.50	4.2	0.49	4.0	0.49
Cobalt	ICAP	20	0.98	9.2	1.0	6.1	0.94	9.5	0.99	2.4	0.97	2.6	0.97
Copper	ICAP	280	0.98	1100	1.0	470	0.94	15000	5.0	700	0.97	680	0.97
Iron	ICAP	61000	49	52000	50	130000	47	82000	50	37000	49	39000	49
Lead	ICAP	560	3.9	880	4.0	20000	3.8	4800	4.0	1700	3.9	1800	3.9
Magnesium	ICAP	6300	49	5500	50	180	47	4900	50	800	49	810	49
Manganese	ICAP	1100	0.98	770	1.0	66	0.94	980	0.99	72	0.97	85	0.97
Mercury	Cold Vapor	0.64	0.04	0.52	0.04	39	0.78	2.1	80.0	1.7	0.04	1.1	0.04
Nickel	ICAP	31	0.98	15	1.0	2.5	0.94	17	0.99	4.0	0.97	3.8	0.97
Potassium	ICAP	2500	200	1600	200	700	190	1600	200	1500	190	1400	190
Selenium	AA-Fur	U	2.4	U	2.5	3.0	2.4	3.2	2.5	U	2.5	U	2.5
Silver	IÇAP	2.8	0.49	12	0.50	71	0.47	71	0.50	16	0.49	15	0.49
Sodium	ICAP :	52	49	U	50	U	47	120	50	63	49	63	49
Thallium	AA-Fur	Ų	2.4	U	2.5	U	2.4	U	2.5	U	2.5	U	2.5
Vanadium	ICAP	19	2.0	18	2.0	3.7	1.9	14	2.0	8.4	1.9	8.4	1.9
Zinc	ICAP	1500	2.0	1600	2.0	11000	9.5	1500	2.0	1300	1.9	920	1.9

Client ID Location		B03 2		B03		B03		B03		B03	616 5	B03	617 4
Parameter	Analysis Method	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg
Aluminum	ICAP	12000	18	670	18	560	17	2300	17	3200	18	3900	17
Antimony	ICAP	54	5.9	1900	5.9	910	5.7	600	5.8	140	5.9	Ü	5.8
Arsenic	ICAP	400	7.4	370	7.4	280	7.1	270	7.2	1200	7.4	1500	7.3
Barium	1CAP	220	0.99	560	0.99	300	0.94	730	0.96	460	0.99	120	0.97
Beryllium	ICAP	1.1	0.50	U	0.50	U	0.47	U	0.48	1.0	0.50	Ų	0.49
Cadmium	ICAP	6.1	0.50	180	0.50	470	0.47	160	0.48	36	0.50	4.8	0.49
Calcium	ICAP	1400	50	240	50	Ų	47	140	48	11000	50	2000	49
Chromium	ICAP	13	0.50	U	0.50	Ų	0.47	3.1	0.48	3.7	0.50	5.6	0.49
Cobalt	ICAP	10	0.99	U	0.99	U	0.94	1.5	0.96	5.4	0.99	1.7	0.97
Соррег	ICAP	2300	0.99	330	0.99	340	0.94	310	0.96	890	0.99	67	0.97
iron	ICAP	59000	50	12000	9.9	15000	9.4	18000	9.6	60000	50	58000	49
Lead	ICAP	3800	4.0	32000	20	21000	3.8	17000	3.8	5200	4.0	340	3.9
Magnesium	ICAP	2600	50	250	50	73	47	680	48	6200	50	490	49
Manganese	IÇAP	340	0.99	82	0.99	460	0.94	120	0.96	280	0.99	20	0.97
Mercury	Cold Vapor	1.7	0.04	51	1.6	67	1.6	35	0.75	4.2	0.19	0.26	0.04
Nickel	ICAP	21	0.99	U	0.99	U	0.94	2.1	0.96	11	0.99	7.8	0.97
Potassium	ICAP	1800	200	650	200	650	190	1100	190	1800	200	6300	190
Selenium	AA-Fur	2.5	2.5	5.0	2.5	4.2	2.4	5.6	2.5	6.4	2.4	Ų	2.5
Silver	ICAP	16	0.50	160	0.50	180	0.47	75	0.48	28	2.5	1.5	0.49
Sodium	ICAP	υ	50	U	50	U	47	U	48	53	50	84	49
Thallium	AA-Fur	U	2.5	U	2.5	Ų	2.4	U	2.5	U	2.4	U	2.5
Vanadium	ICAP	20	2.0	Ų	2.0	2.0	1.9	7.0	1.9	5.2	2.0	7.7	1.9
Zinc	ICAP	1100	2.0	23000	9.9	59000	9.5	21000	9.6	4400	2.0	160	1.9

Client ID Location		B03 5		803 5		803 54 E		
Parameter	Analysis Method	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	
Aluminum	ICAP	11000	18	1900	17	1700	17	
Antimony	ICAP	U	5.9	130	5.8	780	5.8	
Arsenic	ICAP	43	7.4	820	7.2	820	7.3	
Barium	ICAP	86	0.98	110	0.96	110	0.97	
Beryllium	ICAP	1.6	0.49	Ų	0.48	Ų	0.49	
Cadmium	ICAP	1,1	0.49	51	0.48	38	0.49	
Calcium	ICAP	27000	49	1200	48	1300	49	
Chromium	1CAP	18	0.49	2,0	0.48	1.8	0.49	
Cobalt	ICAP	12	0.98	1.9	0.96	2.6	0.97	
Copper	ICAP	50	0.98	760	0.96	1500	0.97	
iron	IÇAP	22000	9.8	97000	48	93000	49	
Lead	ICAP	86	3.9	21000	3.8	25000	3.9	
Magnesium	ICAP	10000	49	610	48	630	49	
Manganese	ICAP	420	0.98	72	0.96	58	0.97	
Mercury	Cold Vapor	0.34	0.04	4.8	0.18	5.4	0.19	
Nickel	ICAP	21	0.98	3.5	0.96	3.6	0.97	
Potassium	ICAP	3500	200	2200	190	2100	190	
Selenium	AA-Fur	U	2.5	3.5	2.4	υ	2.4	
Silver	ICAP	0.69	0.49	64	0.48	92	0.49	
Sodium	ICAP	U	49	U	48	U	49	
Thallium	AA-Fur	U	2.5	2.8	2.4	U	2.4	
Vanadium	ICAP	16	2.0	7.3	1.9	6.9	1.9	
Zinc	ICAP	240	2.0	6700	1.9	5100	1.9	

Client ID Location		Method La		B03i 59		B036		B036		B03		B03 8	
Parameter	Analysis Method	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg
Aluminum	ICAP	U	18	4000	17	850	17	2200	18	1600	18	1600	18
Antimony	ICAP	U	6.0	U	5.8	2800	5.8	1500	5.9	150	5.9	U	5.9
Arsenic	ICAP	U	7.5	230	7.2	560	7.3	520	7.4	160	7.4	81	7.4
Barium	ICAP	U	1.0	160	0.96	5.5	0.97	160	0.99	60	0.98	45	0.99
Beryllium	ICAP	Ų	0.50	1.8	0.48	U	0.49	0.81	0.50	Ų	0.49	U	0.50
Cadmium	ICAP	Ü	0.50	1.4	0.48	220	0.49	210	0.50	47	0.49	U	0.50
Calcium	IÇAP	U	50	49000	48	100000	240	44000	50	1900	49	210	50
Chromium	ICAP	U	0.50	6.9	0.48	3.1	0.49	9.5	0.50	1.6	0.49	1.4	0.50
. Cobalt	ICAP -	U	1.0	-16	0.96	7.2	.0.97	7.7	0.99	U	0.98	U	0.99
Copper	ICAP	U	1.0	36	0.96	3000	0.97	3600	0.99	1600	0.98	37	0.99
Iron	ICAP	U	10	23000	9.6	42000	49	55000	50	61000	49	11000	9.9
Lead	ICAP	IJ	4.0	95	3.8	95000	19	23000	4.0	30000	3.9	930	4.0-
Magnesium	ICAP	U	50	22000	48	56000	49	26000	50	470	49	180	50
Manganese	ICAP	U	1.0	710	0.96	1500	0.97	320	0.99	16	0.98	14	0.99
Mercury	Cold Vapor	U	0.04	0.15	0.05	3.0	0.36	32	0.98	2.1	0.05	0.3	0.05
Nickel	ICAP	Ų	1.0	35	0.96	11	0.97	36	0.99	U	0.98	U	0.99
Potassium	ICAP	IJ	200	2600	190	U	190	1100	200	1300	200	1900	200
Selenium	AA-Fur	U	0.50	U	2.3	4.7	2.3	U	2.5	U	2.3	Ų	2.4
Silver	ICAP	U	0.50	0.64	0.48	430	0.49	69	0.50	120	0.49	6.2	0.50
Sodium	ICAP	U	50	74	48	U	49	76	50	U	49	Ų	50
Thallium	AA-Fur	U	0.50	1.1	0.92	U	0.93	U	0.98	U	0.93	Ú	0.96
Vanadium	ICAP	U	2.0	6.9	1.9	5.0	1.9	10	2.0	3.7	2.0	U	2.0
Zinç	ICAP	U	2.0	190	1.9	29000	9.7	20000	9.9	5700	9.8	46	2.0

Client ID Location		B03 9		B03-		B03 9-	
Parameter	Analysis Method	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg	Conc mg/kg	MDL mg/kg
Aluminum	ICAP	10000	18	760	17	7400	17
Antimony	IÇAP	150	5.9	250	5.6	6.0	5.7
Arsenic	ICAP	300	7.4	660	6.9	560	7.1
Barium	ICAP	800	0.98	510	0.93	220	0.94
Beryllium	ICAP	0.73	0.49	U	0.46	U	0.47
Cadmium	ICAP	20	0.49	100	0.46	2.6	0.47
Calcium	ICAP	14000	49	190	46	170	47
Chromium	ICAP	15	0.49	0.47	0.46	10	0.47
Cobatt	ICAP	5.2	0.98	U	0.93	4.4	0.94
Copper	ICAP	550	0.98	990	0.93	170	0.94
Iron	ICAP	17000	9.8	31000	46	31000	47
Lead	ICAP	11000	3.9	41000	19	310	3.8
Magnesium	ICAP	2700	49	110	46	2400	47 -
Manganese	ICAP	250	0.98	64	0.93	280	0.94
Mercury	Cold Vapor	4.7	0.50	72	2.0	0.42	0.04
Nickel	ICAP	8.0	0.98	U	0.93	9.2	0.94
Polassium	IÇAP	940	200	1600	190	2500	190
Selenium	AA-Fur	U	2.5	U	2.3	U	2.4
Silver	ICAP	49	0.49	140	0.46	1.3	0.47
Sodium	ICAP	1900	49	U	46	150	47
Thallium	AA-Fur	U	0.99	U	0.93	U	0.97
Vanadium	ICAP	18	2.0	2.4	1.9	29	1.9
Zinc	ICAP	2600	2.0	13000	9.3	360	1.9

QA/QC for Metals in Water

Results of the QC Standard Analysis for Metals in Water

QC standards QC-21x100, QC-7x100, ERA-438, TMAA #1, TMAA #2 and SDWA-3034 were used to check the accuracy of the calibration curve. The percent recoveries, listed in Table 2.1, ranged from 83 to 109 and all nineteen recovered concentrations for which 95% confidence limits are available were within these limits. 95% Confidence limits are not available for seventeen values.

Results of the MS/MSD Analysis for Metals in Water

Samples A 04460 and A 04465 were chosen for the matrix spike/matrix spike duplicate analysis (MS/MSD). The percent recoveries, listed in Table 2.2, ranged from 59 to 104 and seventy-two out of seventy-four calculated values were within the acceptable QC limits. Two other values were not calculated because the concentration of analyte in the sample was greater than four times the concentration spiked. The relative percent differences, also listed in Table 2.2, ranged from 0 (zero) to 31 and thirty-six out of thirty-seven calculated values were within the acceptable QC limits. One other value was not calculated because the concentration of analyte in the sample was greater than four times the concentration spiked.

Results of the Blank Spike Analysis for Metals in Water

The results of the blank spike analysis are reported in Table 2.3. The percent recoveries ranged from 93 to 104 and all twenty-three values were within the acceptable QC limits.

Results of the LCS Analysis for Metals in Water

LCS standard 99104 was also analyzed. The percent recoveries, listed in Table 2.4, ranged from 83 to 101 and all nineteen percent recoveries were within the 95% confidence limits.

Table 2.1 Results of the QC Standard Analysis for Metals in Water WA # 0-232 Yankee Mine Site

Metal	Date Analyzed	Quality Control Standard	Conc. Rec µg/L	Certified Value µg/L	95% Confidence Interval µg/L	% Rec
Aluminum	10/25/01	QC-7 x100	1003	1000	NA	100
	10/25/01	ERA-438	587	558	458 - 658	105
Antimony	10/25/01	TMAA#2	59.14	60	38.07 - 69.71	99
Arsenic	11/02/01	TMAA #1	41.46	40	32.45 - 46.55	104
Barium	10/25/01	QC-7 x100	1010	1000	NA	101
	10/25/01	ERA-438	588	583	478 - 688	101
Beryllium	10/25/01	QC-21 x100	1029	1000	NA	103
	10/25/01	ERA-438	99.5	95.8	78.6 - 113	104
Cadmium	10/25/01	QC-21 x100	1024	1000	NA	102
	10/25/01	ERA-438	76.7	75	61.5 - 88.5	102
Calcium	10/25/01	QC-21 x100	1028	1000	NA	103
Chromium	10/25/01	QC-21 x100	1043	1000	NA	104
	10/25/01	ERA-438	542	517	424 - 610	105
Cobatt	10/25/01	QC-21 x100	1061	1000	NA	106
	10/25/01	ERA-438	226	208	171 - 245	109
Copper	10/25/01	QC-21 x100	1022	1000	NA	102
	10/25/01	ERA-438	198	192	157 - 227	103
tron	10/25/01	QC-21 x100	1049	1000	NA	105
	10/25/01	ERA-438	913	867	711 - 1020	105
Lead	10/25/01	TMAA#1	84.1	80	68.98 - 91.16	105
Magnesium	10/25/01	QC-21 x100	963	1000	NA	96
Manganese	10/25/01	QC-21 x100	1044	1000	NA	104
	10/25/01	ERA-438	177	171	140 - 202	104
Mercury	10/26/01	SDWA-3034	3.15	3.8	2.66-4.94	83
Nickel	10/25/01	QC-21 x100	1065	1000	NA	107
	10/25/01	ERA-438	197	187	153 - 221	105
Potassium	10/25/01	QC-7 x100	8961	10000	NA	90
Selenium	10/25/01	TMAA #1	81.1	80	65.38 - 88.47	101
Silver	10/25/01	QC-7 x100	1033	1000	NA	103
	10/25/01	ERA-438	77.1	77.5	63.6 - 91.5	99
Sodium	10/25/01	QC-7 x100	993	1000	NA	99
Thallium	10/24/01	TMAA #2	60.84	60	50.19 - 68.32	101
Vanadium	10/25/01	QC-21 x100	1008	1000	NA	101
	10/25/01	ERA-438	185	192	157 - 227	96
Zinc	10/25/01	QC-21 x100	1034	1000	NA	103
	10/25/01	ERA-438	471	454	372 - 536	104

Table 2.2 Results of the MS/MSD Analysis for Metals in Water WA # 0-232 Yankee Mine Site

Sample ID:	A04460	MS			MSD					
Metal	Sample Conc µg/L	Spike Added µg/L	MS Conc µg/L	MS % Rec	Spike Added µg/L	MSD Conc µg/L	MSD % Rec	RPD	Recomm QC Lit % Rec	
Aluminum	1570	2222	3528	88	2222	3167	72 *	20	75-125	20
Antimony	13.2	55.6	64.6	93	55.6	65.7	95	2	75-125	20
Arsenic	97.7	55.6	141	78	55.6	139	75	5	75-12 5	20
Barium	67.6	222	282	96	222	280	96	1	75-125	20
Beryllium	U	222	210	95	222	210	95	0	75-125	20
Cadmium	7.4	222	207	90	222	207	90	0	75-125	20
Chromium	U	222	211	95	222	210	95	0	75- 125	20
Coball	U	222	209	94	222	207	93	1	75-125	20
Copper	142	222	349	93	222	353	95	2	75-125	20
Iron	13690	2222	15400	NC	2222	15290	NC	NC	75-125	20
Lead	178	55.6	223	81	55.6	21 1	59 •	31	* 75-125	20
Manganese	109	222	314	92	222	312	91	1	75-125	20
Mercury	0.276	2.00	2.29	101	2.00	2.21	97	4	75-125	20
Nickel	12.1	222	213	90	222	215	91	1	75-125	20
Selenium	U	55.6	46.4	84	55.6	46.2	83	0	75-125	20
Silver	Ų	222	198	89	222	201	90	2	75-125	20
Thallium	U	55.6	57.4	103	55.6	57.8	104	1	75-125	20
Vanadium	U	222	211	95	222	213	96	1	75-125	20
Žinc	1013	222	1200	84	222	1187	78	7	7 5- 125	20

Table 2.2 (cont.) Results of the MS/MSD Analysis for Metals in Water WA # 0-232 Yankee Mine Site

Sample ID:	A04465 Sample	MS Spike	MS	MS	MSD Spike	MSD	MSD		Recomm	hehner
Metal	Conc µg/L	Added µg/L	Conc µg/L	% Rec	Added µg/L	Conc µg/L	% Rec	RPD	QC Li % Rec	mits
Aluminum	103	2222	2184	94	2222	2208	95	1	75-125	20
Antimony	U	55,6	55.9	101	55.6	57.2	103	2	75-125	20
Arsenic	U	55.6	53.4	96	55.6	53.3	96	0	75-125	20
Barium	65.7	222	279	96	222	282	97	1	75-125	20
Beryllium	U	222	209	94	222	211	95	1	75-125	20
Cadmium	5	222	209	92	222	207	91	1	75-125	20
Chromium	U	222	213	96	222	214	96	0	75-125	20
Cobalt	U	222	212	95	222	213	96	0	75-125	20
Copper	62.9	222	275	95	222	280	98	2	75-125	20
Iron	99.3	2222	2221	95	2222	2247	97	1	75-125	20
Lead	10.8	55.6	59.7	88	55.6	59.2	87	1	75-125	20
Manganese	84.7	222	292	93	222	295	95	1	75-125	20
Mercury	U	2.00	1.87	94	2.00	1.91	96	2	75-125	20
Nicket	U	222	222	100	222	216	97	3	75-125	20
Selenium	U	55.6	46	83	55.6	45	81	2	75-125	20
Silver	U	222	203	91	222	205	92	1	75-125	20
Thallium	U	55.6	56.4	102	55.6	55.1	99	2	75-125	20
Vanadium	U	222	215	97	222	216	97	0	75-125	20
Zinc	765	222	960	88	222	967	91	,4	75-125	20

Table 2.3 Results of the Blank Spike Analysis for Metals in Water
WA # 0-232 Yankee Mine Site

Metal	Spiked Conc. µg/L	Rec Conc. µg/L	% Rec	Recommended QC Limits %Rec
Aluminum	2222	2075	93	75-125
Antimony	55. 6	55.3	100	75-125
Arsenic	55.6	54.7	98	75-125
Barium	222	213	96	75-125
Beryllium	222	211	95	75-125
Cadmium	222	207	93	75-125
Calcium	2222	2108	95	75-125
Chromium	222	218	98	75-125
Cobalt	222	218	98	75-125
Copper	222	214	96	75-125
tron	2222	2174	98	75-125
Lead	55. 6	56	101	75-12 5
Magnesium	2222	2057	93	75-125
Manganese	222	214	96	75-125
Mercury	2.00	1.96	98	75-125
Nickel	222	219	99	75-125
Potassium	8889	8407	95	75-125
Selenium	55.6	57.6	104	7 5-12 5
Silver	222	208	94	75-125
Sodium	2222	2085	94	75-125
Thallium	55. 6	54	97	75-125
Vanadium	222	216	97	75-125
Zinc	222	209	94	75-125

Table 2.4 Results of the LCS Analysis for Metals in Water WA # 0-232 Yankee Mine Site

Metal	Date Analyzed	LCS Standard (ERA Lot#)	Conc. Rec µg/L	Certified Value µg/L	PALs µg/L	% Rec
Aluminum	10/25/01	99104	1077	1120	918 - 1320	96
Antimony	10/25/01	99104	175	183	137 - 229	96
Arsenic	11/02/01	99104	281	295	221 - 348	95
Barium	10/25/01	99104	959	975	800 - 1150	98
Beryllium	10/25/01	99104	560	572	469 - 675	98
Cadmium	10/25/01	99104	455	508	417 - 599	90
Chromium	10/25/01	99104	889	902	740 - 1060	99
Cobalt	10/25/01	99104	237	240	197 - 283	99
Copper	10/25/01	99104	634	643	527 - 759	99
iron	10/25/01	99104	428	445	365 - 525	96
Lead	10/25/01	99104	298	296	243 - 349	101
Manganese	10/25/01	99104	1482	1520	1250 - 1790	98
Mercury	10/26/01	99104	13.5	16.3	12.2 - 20.4	83
Nickel	10/25/01	99104	1755	1790	1470 - 2210	98
Selenium	10/25/01	99104	1250	1400	1050 - 1650	89
Silver	10/25/01	99104	541	578	474 - 682	94
Thallium	10/24/01	99104	573	610	458 - 720	94
Vanadium	10/25/01	99104	1395	1420	1160 - 1680	98
Zinc	10/25/01	99104	788	843	691 - 995	93

QA/QC for Metals in Soil

Results of the QC Standard Analysis for Metals in Soil

QC standards QC-21x100, QC-7x100, ERA-438, TMAA #1, TMAA #2 and SDWA-3034 were used to check the accuracy of the calibration curve. The percent recoveries, listed in Table 2.5, ranged from 97 to 111 and all thirty-six recovered concentrations for which 95% confidence are available were within these limits. 95% Confidence limits are not available for thirty-eight values.

Results of the MS/MSD Analysis for Metals in Soil

Samples B 03607, B 03616 and B 03625 were chosen for the matrix spike/matrix spike duplicate analysis (MS/MSD). The percent recoveries, listed in Table 2.6, ranged from 1 to 632 and sixty-six out of seventy-six calculated values were within the acceptable QC limits. Twenty-six other values were not calculated because the concentration of analyte in the sample was greater than four times the concentration spiked. The relative percent differences, also listed in Table 2.6, ranged from 0 (zero) to 148 and thirty-one out of thirty-eight calculated values were within the acceptable QC limits. Thirteen other values were not calculated because the concentration of analyte in the sample was greater than four times the concentration spiked.

Results of the Blank Spike Analysis for Metals in Soil

The results of the blank spike analysis are reported in Table 2.7. The percent recoveries ranged from 86 to 112 and all forty-six values were within the acceptable QC limits.

Results of the LCS Analysis for Metals in Soil

LCS standard 248 was also analyzed. The percent recoveries, listed in Table 2.8, ranged from 40 to 108 and all forty-six percent recoveries were within the 95% confidence limits.

Table 2.5 Results of the QC Standard Analysis for Metals in Soil WA # 0-232 Yankee Mine Site

Metal	Date Analyzed	Quality Control Standard	Conc. Rec µg/L	Certified Value µg/L	95% Confidence Interval µg/L	% Rec
Aluminum	10/18/01	QC-7 x100	1020	1000	NA	102
	10/18/01	ERA-438	610	558	458 - 656	109
Antimony	10/18/01	QC-21 x100	1027	1000	NA	103
Arsenic	10/18/01	QC-21-x100	1014	194000-	es e NA s est	101-1-
Barium	10/18/01	QC-7 x100	1005	1000	NA	100
	10/18/01	ERA-438	581	583	478 - 6 88	100
Beryllium	10/18/01	QC-21 x100	1038	1000	NA	104
	10/18/01	ERA-438	100	95.8	78.6 - 113	104
Cadmium	10/18/01	QC-21 x100	1020	1 00 0	NA	102
	10/18/01	ERA-438	78	75	61.5 - 88.5	104
Calcium	10/18/01	QC-21 x100	1028	1000	NA	103
Chromium	10/18/01	QC-21 x100	1053	1000	NA	105
	10/18/01	ERA-438	545	517	424 - 610	105
Cobalt	10/18/01	QC-21 x100	1072	1000	NA	107
	10/18/01	ERA-438	231	208	171 - 245	111
Copper	10/18/01	QC-21 x100	1033	1000	NA	103
	10/18/01	ERA-438	198	192	157 - 227	103
Iron	10/18/01	QC-21 x100	1083	1000	NA	108
	10/18/01	ERA-438	943	867	711 - 1020	109
Lead	10/18/01	QC-21 x100	1053	1000	NA	105
	10/18/01	ERA-438	308	292	239 - 345	105
Magnesium	10/18/01	QC-21 x100	996	1000	NA	100
Manganese	10/18/01	QC-21 x100	1053	1000	NA	105
	10/18/01	ERA-438	179	171	140 - 202	105
Mercury	10/10/01	SDWA-3042	3.13	3.18	2.23 - 4.13	98
Nickel	10/18/01	QC-21 x100	1087	1000	NA	109
	10/18/01	ERA-438	201	187	153 - 221	107
Potassium	10/18/01	QC-7 x100	10020	10000	NA	100
Selenium	10/24/01	TMAA #1	80.2	80	65.38 - 88.47	100
Silver	10/18/01	QC-7 x100	1032	1000	NA	103
	10/18/01	ERA-438	80	77.5	63.6 - 91.5	103
Sodium	10/18/01	QC-7 x100	1010	1000	NA	101
Thallium	10/24/01	TMAA #2	64.09	60	50.19 - 68.32	107
Vanadium	10/18/01	QC-21 x100	1033	1000	NA	103
	10/18/01	ERA-438	193	192	157 - 227	101
Zinc	10/18/01	QC-21 x100	1040	1000	NA	104
	10/18/01	ERA-438	475	454	372 - 536	105

Table 2.5 (cont.) Results of the QC Standard Analysis for Metals (Soil) WA # 0-232 Yankee Mine Site

UMAN STREET, I HAVE UP SPECIAL STREET

Metal	Date Analyzed	Quality Control Standard	Conc. Rec µg/L	Certified Value µg/L	95% Confidence Interval µg/L	% Rec
Aluminum	10/17/01	QC-7 x100	1019	1000	NA	102
	10/17/01	ERA-438	618	558	458 - 658	111
Antimony	10/17/01	QC-21 x100	1012	1000	81.7 - 125	101
Arsenic	10/17/01	QC-21 x100	1034	1000	81.7 - 125	103
Banum	10/17/01	QC-7 x100 ERA-438	1011 581	1000 583	NA 478 - 688	101 100
Beryllium	10/17/01	QC-21 x100	1030	1000	NA	103
	10/17/01	ERA-438	99	95.8	78.6 - 113	103
Cadmium	10/17/01	QC-21 x100	1018	1000	NA	102
	10/17/01	ERA-438	79	75	61.5 - 88.5	105
Calcium	10/17/01	QC-21 x100	1028	1000	NA	103
Chromium	10/17/01	QC-21 x100	1049	1000	NA	105
	10/17/01	ERA-438	547	51 7	424 - 610	106
Cobalt	10/17/01	QC-21 x100	1075	1000	NA	108
	10/17/01	ERA-438	231	208	171 - 245	111
Copper	10/17/01	QC-21 x100	1025	1000	NA	102
	10/17/01	ERA-438	199	192	157 - 227	104
Iron	10/17/01	QC-21 x100	1070	1000	NA	107
	10/17/01	ERA-438	935	867	711 - 1020	108
Lead	10/17/01	QC-21 x100	1049	1000	NA	105
	10/17/01	ERA-438	316	292	239 - 345	108
Magnesium	10/17/01	QC-21 x100	987	1000	NA	99
Manganese	10/17/01	QC-21 x100	1049	1000	NA	105
	10/17/01	ERA-438	177	171	140 - 202	104
Mercury	10/12/01	SDWA-3042	3.08	3.18	2.23 - 4.13	97
Nickel	10/17/01	QC-21 x100	1073	1000	NA	107
	10/17/01	ERA-438	205	187	153 - 221	110
Potassium	10/17/01	QC-7 x100	9816	10000	NA	98
Selenium	10/23/01	TMAA #1	81.5	80	65.38 - 88.47	102
Silver	10/17/01	QC-7 x100	1025	1000	NA	102
	10/17/01	ERA-438	81	77.5	63.6 - 91.5	105
Sodium	10/17/01	QC-7 x100	1019	1000	NA	102
Thallium	10/23/01	TMAA #2	63.36	60	50.19 - 68.32	106
Vanadium	10/17/01	QC-21 x100	1025	1000	NA	102
	10/17/01	ERA-438	195	192	157 - 227	102
Zinc	10/17/01	QC-21 x100	1034	1000	NA	103
	10/17/01	ERA-438	469	454	372 - 536	103

Table 2.6 Results of the MS/MSD Analysis for Metals in Soil WA # 0-232 Yankee Mine Site Results based on as received weight

Sample ID:	B03607 Sample Conc	MS Spike Added	MS Conc	MS %	MSD Spike Added	MSD Conc	MSD %		Recomn QC Li	
Metal	mg/kg	mg/kg	mg/kg	Rec	mg/kg	mg/kg	Rec	RPD	%Rec	RPD
Antimony	41.9	49.5	52.6	22 *	48.1	42.2	1 •	98	* 75-125	20
Arsenic-ICAP	287	49.5	297	NC	48.1	302	, NC	NC	75-125	20
Barium	150	49.5	207	115	48.1	293	297 *	88	* 75-125	20
Beryllium	8.0	49.5	49.3	98	48.1	47.7	98	0	7 5- 125	20
Cadmium	13	49.5	52.7	80	48.1	50	77	4	75-125	20
Chromium	11.4	49.5	59	96	48.1	56	93	4	75-125	20
Cobalt	9.16	49.5	55.2	93	48.1	53.2	92	2	75-125	20
Copper	1053	49.5	918	NC	48.1	1241	NC	NC	75-125	20
Lead	884	49.5	903	NC	48.1	1141	NC	NC	75-125	20
Manganese	772	49.5	813	NC	48.1	821	NC	NÇ	75-125	20
Mercury	0.518	0.400	0.964	111	0.400	0.894	94	17	75-125	20
Nickel	14.6	49.5	61	94	48.1	58.5	91	3	75-125	20
Selenium	Ų	5.00	6.35	127	4.95	6.04	122	4	75-125	20
Silver	11.7	49.5	57.1	92	48.1	51.6	83	10	75-125	20
Thallium	U	5.00	4.86	97	4.95	4.94	100	3	75-125	20
Vanadium	17.7	49.5	63.6	93	48.1	61.5	91	2	75-125	20
Zinc	1627	49.5	1039	NC	48.1	936	NC	NC	75-125	20

Table 2.6 (cont.) Results of the MS/MSD Analysis for Metals in Soil WA # 0-232 Yankee Mine Site Results based on as received weight

Sample ID:	B03616	MS			MSD					
Metal	Sample Conc mg/kg	Spike Added mg/kg	MS Conc mg/kg	MS % Rec	Spike Added mg/kg	MSD Conc mg/kg	MSD % Rec	RPD	Recomn QC Li %Rec	
Antimony	136	49.0	195	120	48.1	154	37 •	105	* 75-125	20
Arsenic-ICAP	1199	49.0	1218	NC	48.1	1171	NC	NC	75-125	20
Barium	458	49.0	449	NC	48.1	426	NC NC	NC	75-125	20
Beryllium	1	49.0	49.6	99	48.1	48.8	99	0	75-125	20
Cadmium	36.4	49.0	77.6	84	48.1	77	84	0	75-125	20
Chromium	3.7	49.0	51.2	97	48.1	50.3	97	0	75-125	20
Cobalt	5.38	49.0	50.1	91	48.1	51.4	96	5	75-125	20
Copper	890	49.0	945	NC	48.1	1314	NC	NC	75-125	20
Lead	5211	49.0	10520	NC	48.1	5211	NC	NC	75-125	20
Manganese	284	49.0	292	NC	48.1	343	NC	NC	75-125	20
Mercury	4.21	0.351	4.39	NC	0.364	4.45	NC	NC	75-125	20
Nickel	10.9	49.0	55.8	92	48.1	56.8	95	4	75-125	20
Selenium	6.36	4.95	7.38	21 '	4.85	6.99	13 *	47	* 75-125	20
Silver	25.5	49.0	79.3	110	48.1	65.9	84	27	* 75-125	20
Thallium	υ	4.95	4.57	92	4.85	4.25	88	5	75-125	20
Vanadium	5.24	49.0	51.1	94	48.1	50.1	93	0	75-125	20
Zinc	4373	49.0	4034	NC	48.1	4376	NC	NC	75-125	20
Selenium Silver Thallium Vanadium	6.36 25.5 U 5.24	4.95 49.0 4.95 49.0	7.38 79.3 4.57 51.1	21 1 110 92 94	4.85 48.1 4.85 48.1	6.99 65.9 4.25 50.1	13 * 84 88 93	47 27 5	* 75-125 * 75-125 75-125 75-125	20 20 20 20

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Table 2.6 (cont.) Results of the MS/MSD Analysis for Metals in Soil WA # 0-232 Yankee Mine Site Results based on as received weight

Sample ID:	B03625	MS			MSD				_	
Metal	Sample Conc mg/kg	Spike Added mg/kg	MS Conc mg/kg	MS % Rec	Spike Added mg/kg	MSD Conc mg/kg	MSD % Rec	RPD	Recomm QC Li %Rec	mits
Antimony	υ	48.5	11	23 *	49.0	13	27 •	16	75-125	20
Arsenic-ICAP	80.7	48.5	129	99	49.0	119	78	24	75-125	20
Barium	45.4	48.5	105	123	49.0	102	115	6	75-125	20
Beryllium	υ	48.5	50.1	103	49.0	48.7	99	4	75-125	20
Cadmium	U	48.5	45.6	94	49.0	42.7	87	8	75-125	20
Chromium	1.38	48.5	50.3	101	49.0	48.5	96	5	75-125	20
Cobalt	U	48.5	47	97	49.0	45.7	93	4	75-125	20
Соррег	37	48.5	84.1	96	49.0	82.4	92	5	75-125	20
Lead	927	48.5	906	NC	49.0	3294	NC	NC	75-125	20
Manganese	14	48.5	62.5	100	49.0	60.2	94	6	75-125	20
Mercury	0.303	0.455	0.795	108	0.488	0.837	109	1	75-125	20
Nickel	Ų	48.5	47.5	98	49.0	45.9	94	4	75-125	20
Selenium	U	4.95	4.26	86	4.90	4.51	92	7	75-125	20
Silver	U	48.5	43.5	90	49.0	40.8	83	7	75-125	20
Thallium	υ	4.95	4.61	93	4.90	4.58	93	0	75-125	20
Vanadium	U	48.5	46.2	95	49.0	44.7	91	4	75-125	20
Zinc	46.2	48.5	351	632	49.0	92.3	94	148	* 75-125	20

Table 2.7 Results of the Blank Spike Analysis for Metals in Soil WA # 0-232 Yankee Mine Site

Metal	Spiked Conc. mg/kg	Blank Conc. mg/kg	Rec Conc. mg/kg	% Recovery	Recommended QC Limits %Rec
Aluminum	400	υ	448	112	75-125
Antimony	50.0	U	48.3	97	75-125
Arsenic	50.0	υ	49.4	98	75-125
Barium	50.0	U	49.1	98	75-125
Beryllium	50.0	U	50.4	101	75-125
Cadmium	50.0	U	46.7	9 3	75-125
Calcium	400	U	383	· 96	75-125
Chromium	50.0	U	49.4	99	75-125
Cobalt	50.0	υ	49.5	99	75-125
Copper	50.0	U	49	98	75-125
Iron	400	U	408	102	75-125
Lead	50.0	U	48.8	98	75-125
Magnesium	400	U	384	96	75-125
Manganese	50.0	U	48.8	98	75-125
Mercury	0.392	U	0.4	102	75-125
Nickel	50.0	U	50.3	101	75-125
Potassium	800	U	711	. 89	75-125
Selenium	4.95	υ	4.99	101	75-125
Silver	50.0	υ	46.4	93	75-125
Sodium	400	U	387	97	75-125
Thallium	4.95	U	5.15	104	75-125
Vanadium	50.0	U	48.6	97	75-125
Zinc	50.0	U	48	96	75-125

Metal	Spiked Conc. mg/kg	Blank Conc. mg/kg	Rec Conc. mg/kg	% Recovery	Recommended QC Limits %Rec
Aluminum	400	υ	421	105	75-125
Antimony	50.0	U	44.7	89	75-125
Arsenic-ICAP	50.0	U	47	94	75-125
Barium	50.0	U	47.7	95	75-125
Beryllium	50.0	U	49.9	100	75-125
Cadmium	50.0	υ	45.4	, 91	75-125
Calcium	400	U	376	94	75-125
Chromium	50.0	U	48.3	97	75-125
Cobalt	50.0	U	48.3	97	75-125
Copper	50.0	υ	48.1	96	75-125
Iron	400	U	396	9 9	75-125
Lead	50.0	U M	45	955, milit i i 1 90 0 (^{\$45,675})	75-125
Magnesium	400	U	372	93	75-125
Manganese	50.0	U	47.9	96	75-125
Mercuty	0.400	U	0.4	100	75-125
Nickel	50.0	υ	49	98	75-125
Potassium	800	υ	689	86	75-125
Selenium	5.00	υ	5.03	101	75-125
Silver	50.0	U	45.1	90	75-125
Sodium	400	U	378	95	75-125
Thallium	5.00	U	5.31	106	75-125
Vanadium	50.0	υ	47.4	95	75-125
Zinc	50.0	U	46.7	93	75-125

والأواج فالمتلفظ فليها والمارا والماران والمنافية والمنافية والمنافئة والمتابعة والمتابعة والمتابعة والمتابعة

Table 2.8 Results of the LCS Analysis for Metals in Soil WA # 0-232 Yankee Mine Site

Metal	Date Analyzed	LCS Standard (ERA Lot#)	Conc. Rec µg/L	Certified Value µg/L	PALs µg/L	% Rec
Aluminum	10/18/01	LCS 248	8722	9200	5300 - 13100	95
Antimony	10/18/01	LCS 248	26	62.7	17.1 - 141	41
Arsenic	10/18/01	LCS 248	51	47.5	34.4 - 60.6	107
Barium	10/18/01	LCS 248	505	509	392 - 626	99
Beryllium	10/18/01	LCS 248	59.9	55.9	43.8 - 68.2	107
Calcium	10/18/01	LCS 248	11794	11700	8740 - 14600	101
Cadmium	10/18/01	LCS 248	162	157	118 - 196	103
Chromium	10/18/01	LCS 248	51.4	51.4	39.0 - 63.7	100
Cobalt	10/18/01	LCS 248	91	88.4	68.8 - 108	103
Соррег	10/18/01	LCS 248	71.3	69.5	56.9 - 82.0	103
Iron	10/18/01	LCS 248	13020	13700	8350 - 19100	95
Lead	10/18/01	LCS 248	184	186	139 - 233	99
Magnesium	10/18/01	LCS 248	3041	3070	2280 - 3860	99
Manganese	10/18/01	LCS 248	679	674	511 - 836	101
Mercury	10/10/01	LCS 248	6.42	6.21	4.19 - 8.23	103
Nickel	10/18/01	LCS 248	117	112	87.6 - 137	104
Potassium	10/18/01	LCS 248	3306	3640	2670 - 4610	91
Selenium	10/24/01	LCS 248	115	107	66.3 - 148	107
Silver	10/18/01	LCS 248	85.6	84.3	54.2 - 114	102
Sodium	10/18/01	LCS 248	807	863	585 - 1140	94
Thallium	10/24/01	LCS 248	67. 5	68.1	39.0 - 97.4	99
Vanadium	10/18/01	LCS 248	134	136	92.6 - 179	99
Zinc	10/18/01	LCS 248	268	289	224 - 356	100

Table 2.8 (cont.) Results of the LCS Analysis for Metals in Soil WA # 0-232 Yankee Mine Site

THE THEORY OF THE PROPERTY OF

	Analyzed	LCS Standard	Conc. Rec	Certified Value	PALs	% Rec
		(ERA Lot#)	µg/L	µg/L	µ9/L	
Numinum	10/17/01	LCS 248	7821	9200	5300 - 13100	85
Antimony	10/17/01	LCS 248	25.1	62.7	17.1 - 141	40
Arsenic-ICAF	10/17/01	LCS 248	47.5	47.5	34.4 - 60.6	100
Barium .	10/17/01	LCS 248	486	509	392 - 626	95
Beryllium	10/17/01	LCS 248	60.4	55.9	43.8 - 68.2	108
Calcium	10/17/01	LCS 248	12059	11700	8740 - 14600	103
Cadmium	10/17/01	LCS 248	163	157	118 - 196	104
Chromium	10/17/01	LCS 248	51.3	51.4	39.0 - 63.7	100
Cobalt	10/17/01	LCS 248	90.7	88.4	68.8 - 108	103
Copper	10/17/01	LCS 248	72.8	69.5	56.9 - 82.0	105
ron	10/17/01	LCS 248	12703	13700	8350 - 19100	93
_ead	10/17/01	LCS 248	192	186	139 - 233	103
Magnesium	10/17/01	LCS 248	2944	3070	2280 - 3860	96
Manganese	10/17/01	LCS 248	689	674	511 - 836	102
Mercury	10/12/01	LCS 248	5.93	6.21	4.19 - 8.23	95
Nickel	10/17/01	LCS 248	117	112	87.6 - 137	104
Potassium	10/17/01	LCS 248	3146	3640	2670 - 4610	86
Selenium	10/23/01	LCS 248	115	107	66.3 - 148	107
Silver	10/17/01	LCS 248	85	84.3	54.2 - 114	101
Sodium	10/17/01	LCS 248	776	863	585 - 1140	90
Thailium	10/23/01	LCS 248	72.3	68.1	39.0 - 97.4	106
Vanadium	10/17/01	LCS 248	131	136	92.6 - 179	96
Zinc	10/17/01	LCS 248	293	289	224 - 356	101

REAC, Edison, N (732) 321-4200 EPA Contract 68-C99-223 CHAIN OF CUSTODY RECORD

Project Name: Xantee Mine Site
Project Number: RIADO332

LM Contact: 6ROSIMAN Phone: (732)321-4230

00537 Sheet 01 of 01(Do not copy) (for addnt. samples use new form)

100901 Sample Identification **Analyses Requested** TAL METALS Sample No Matrix Date Collected # of Bottles REAC# Sampling Location Container/Preservative B03620 492 54 DUP 1. 2017 2001 XRF (UP /4.C B03621 59 30052001 B03622 994 64 995 B03623 803624 75 496 ¥ B03625 997 86 B03626 97 997 4 OCT 2001 B03627 999 3 OCT 2011 9-41 >1000 303628

Mairix:

A- Air AT-Animal Tissue DL- Drum Liquids

DS- Drum Solids GW- Groundwater O- Oil

PR-Product PT-Plant Tissue PW- Potable Water QA/QC.

S- Soil SD- Sediment SL- Sludge

8W- Surface Water W- Water X- Other

Special Instructions:

* MU/MUD

SAMPLES TRANSFERRED FROM

CHAIN OF CUSTODY #:

items/Remon	Relinguished by	Date	Received by	Date	Time	ltems/Reason	Relinquished by	Date	Received by	Date,	Time
ALL/ANALYSO	M	11/-1/2	1 drus	U-9-01	6000	911/Tel	myst	10-9-01	Coasser	10/9/01	16:10
(V 11				116			17	
						IL)	

REAC, Edison, NJ (732) 321-4200 EPA Contract 68-C99-223

CHAIN OF CUSTODY RECORD

Project Name: Yunkee Mine Site Project Number: RIA0932

Phone: (133) 321-4230 LM Contact: (RONMAN

00536 No: Sheet 01 of 01(Do not copy) (for addnl. samples use new form)

10401-

QA/QC:

		Sample Identificat	ion			Analyses Requested						
REAC#	Sample No	Sampling Location	Matrix	Date Collect	d # of Bottles	Container/Preservative	TAL METALS					
973	B03601	a	<u>S_</u>	10000	01 1	YRF CUP / 4°C	√					
974	B03602	4				<u> </u>						
975	B03603	チ		}		<u> </u>						
976	B03604	10										
977	B03605	10 DUP							<u> </u>			
978	BAZLAG	12										
979	B03607 *	14										
980	B03608	18										
781	B03609	20		₩			1 1					
792	B03610	23		2007 20	01					$\Lambda_{}$		
983	803611	23 DOP										
484	1303612	25				<u> </u>						
925	1303613	28		<u> </u>		<u> </u>						
986	B03614	29							/			
987	B03615	29 32	<u> </u>						<u> </u>			
928	B03616	35										
939	B03617	44										
990	B03618	50										
991	B03619	54				<u> </u>		/				

M	Ltr	İΧ	
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A- Air AT-Animal Tissue

DL- Drum Liquids DS- Drum Solids GW- Groundwater

O- Oil PR-Product PT-Plant Tissue

SW-Smilete Wale TX-TCLP Extract W- Water X- Other

SD: Sediment

SL- Sludge

S- Soil

PW- Potable Water

Special Instructions:

* MS/MSP

9-7-6 open

SAMPLES TRANSFERRED FROM

CHAIN OF CUSTODY #:

ACCIONALISIS PAR 10/4/17 JULY # 10-9-01 OPEN GIL/TAL JULE 10-9-01 Ch	asses 10/9/01	16:10
	1/2	
		_1

REAC, Edison, NJ (732) 321-4200 EPA Contract 68-C99-223

Project Name: Yankee Mine Sife
Project Number: KIAGOZZZ

Phone: (732) 321-4230 LM Contact: 6 QUISMAN

No:

00358

Sheet 01 of 01(Do not copy) (for addni. samples use new form)

100901-

Samula Identification

Analyses Degreeted

100901-		Sample Identification				Analyses Requested							
REACH	Sanapale No	Sampling Location	Matrix	Date Collec	ted # a	of Bottles	Container/Preservative	TAL MEIALS					
957	A See A 0 4451	SW-01	W.	30(120	el	:1	IL Puly /4.6 HNO3	v				7.	
958	A04452	SW-01 DOP		<u> </u>		<u> </u>						<i></i>	
959	A 04453	5w-02											
960	A04454	ათ-იშ									/		
961	A04455	5 W - 04										-	
962	A0-1456	5 w = 0.5											
963	F 2 1 1 0 9	Sw-06						<u> </u>					
964	A 04458	Sw-c7	<u> </u>			\bot		<u> </u>					
965	A04459	Sw-08						<u> </u>					
966	AOHYBO	SW-09						<u> </u>	<u> </u>				
967	AOYYEL	5w-10		<u> </u>	_								
968	A04462	S (4 - 1)				_ _							
969	A04463	S (2 - 12						<u> </u>	<u> </u>				
970	ACHUBU	566 P1						<u> </u>		/			
971	MO4465	SEEP Z	<u> </u>			_↓							
972	AO446	FIELD BLANK	1	1		3	<u> </u>						
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				ļ					/				
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Matrix:

A- Air

PW- Potable Water

AT-Animal Tissue DL- Drum Liquids S- Soil

DS- Drum Solids

SD- Sediment SL. Sludge

GW- Groundwater O-Oil

SW- Surface Water TX-TCLP Extract

PR-Product PT-Plant Tissue W- Water X- Other

Special Instructions:



SAMPLES TRANSFERRED FROM

CHAIN OF CUSTODY #:

ALL / ANALYSIS	Relinquished by	Date 10 4 0	Received by	Date is 4-1	Time	Items/Reason a 11/6 wl-75'5	Relinguished by	Date	Received by	Date /0/9/0/	Time 15:50
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										^	